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FACULTY OF NATURAL AND ENVIRONMENTAL SCIENCES

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**GEOLOGICAL STORAGE OF CARBON DIOXIDE  
IN OCEANIC CRUST**

by

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Thesis for the degree of Doctor of Philosophy

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## **ABSTRACT**

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Chiara Marieni

The rise of atmospheric carbon dioxide (CO<sub>2</sub>), due to decades of burning of fossil fuels, is a key driver of anthropogenic climate change. Carbon Capture and Storage (CCS) is one of the most promising mitigation strategies for long-term sequestration of CO<sub>2</sub>.

Unlike most conventional CCS investigations targeting deep saline aquifers, this thesis focuses on the potential of the uppermost oceanic crust, inspired by the strong evidence that basaltic seafloor has acted, in the past, as a major sink for CO<sub>2</sub>.

The study of temperature, pressure, and density of CO<sub>2</sub> and seawater at the sediment-basement interface for the whole seafloor highlights the influence of water depth, sediment thickness, and oceanic crustal age on the relative gravitational stability of CO<sub>2</sub>. Consequently, 8% of the entire oceanic crust is recognised as suitable for gravitational and physical trapping of CO<sub>2</sub> injected into the basement. Five potential targets are proposed, and even the smallest of these provides sufficient carbon dioxide sequestration capacity for the next centuries.

Batch experiments on the mineral dissolution of submarine mafic rocks and ophiolitic gabbro, in CO<sub>2</sub>-rich solutions, contribute to improve the fundamental understanding of geochemical reactions at mid-ocean ridge flank temperatures (40 °C). Concentrations of silicon and calcium in solution, and particle size are identified as the key factors to quantify the rock reactivity. Ca dissolution rates suggest calcite, plagioclase and amphibole are the principal sources of calcium at pH ~5.

The attempted estimation of costs related to the transport and storage of 20 Mt/yr of CO<sub>2</sub> in deep-sea basalts, as a function of distance from the shore, injection rate, and water depth, shows the economic feasibility of potential offshore CCS projects. Overall, the expenditures are dominated by the number of ships and wells required to deliver large volumes of CO<sub>2</sub> to reservoirs located far from the coast, rather than by the water depth. These financial considerations could potentially improve if the CCS strategies conquered a significant place in the global market.





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# DECLARATION OF AUTHORSHIP

I, **CHIARA MARIENI**

declare that this thesis and the work presented in it are my own and has been generated by me as the result of my own original research.

## **GEOLOGICAL STORAGE OF CARBON DIOXIDE IN OCEANIC CRUST**

I confirm that:

1. This work was done wholly or mainly while in candidature for a research degree at this University;
2. Where any part of this thesis has previously been submitted for a degree or any other qualification at this University or any other institution, this has been clearly stated;
3. Where I have consulted the published work of others, this is always clearly attributed;
4. Where I have quoted from the work of others, the source is always given. With the exception of such quotations, this thesis is entirely my own work;
5. I have acknowledged all main sources of help;
6. Where the thesis is based on work done by myself jointly with others, I have made clear exactly what was done by others and what I have contributed myself;
7. Parts of this work have been published as: *Marieni et al. [2013]*.

Signed: .....

Date:.....



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## A-Chapter 3

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## A-Chapter 4

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## Abbreviations

|                  |                                                                   |
|------------------|-------------------------------------------------------------------|
| A <sub>BET</sub> | BET surface area                                                  |
| Ac               | carbon Alkalinity                                                 |
| A <sub>geo</sub> | geometric surface area                                            |
| B€               | Billions of euro (x10 <sup>9</sup> )                              |
| barg             | bar gouge (bar-1)                                                 |
| BET              | Brunauer-Emmett-Teller (surface area analysis)                    |
| CAPEX            | CAPital EXpenditure                                               |
| CCS              | Carbon Capture and Storage                                        |
| CDR              | Carbon Dioxide Removal                                            |
| CO <sub>2</sub>  | carbon dioxide                                                    |
| COP21            | Conference Of Parties – 21st                                      |
| DIC              | Dissolved Inorganic Carbon                                        |
| DSDP             | Deep Sea Drilling Project                                         |
| eePO             | eastern equatorial Pacific Ocean                                  |
| EOR              | Enhanced Oil Recovery                                             |
| EPR              | East Pacific Rise                                                 |
| FLNG             | Floating Liquefied Natural Gas                                    |
| FPSO             | Floating Production Storage and Offloading system                 |
| GDH1             | Global Depth and Heat flow model -1                               |
| GEBCO            | GEneral Bathymetric Chart of the Ocean                            |
| GMT              | Generic Mapping Tool                                              |
| GSA              | Geometric Surface Area analysis                                   |
| GSD              | Grain Size Distribution                                           |
| Gt               | Giga tons (x10 <sup>9</sup> )                                     |
| HSCM             | Half Space Cooling Model                                          |
| IAPSO            | International Association for the Physical Sciences of the Oceans |
| IC               | Ion Chromatography                                                |
| ICP-MS           | Inductively Coupled Plasma-Mass Spectrometry                      |
| ICP-OES          | Inductively Coupled Plasma-Optical Emission Spectrometry          |
| IEA GHG          | International Energy Agency Greenhouse Gases                      |
| IODP-1           | Integrated Ocean Drilling Program                                 |
| IODP-2           | International Ocean Discovery Program                             |
| IPCC             | Intergovernmental Panel on Climate Change                         |
| JdF              | Juan de Fuca                                                      |
| K <sub>s</sub>   | thermal conductivity of sediments                                 |
| LNG              | Liquid Natural Gas                                                |
| LOI              | Loss Of Ignition                                                  |
| M€               | Millions of euro (x10 <sup>6</sup> )                              |
| Ma               | Millions of year ago (x10 <sup>6</sup> )                          |
| MAR              | Mid-Atlantic Ridge                                                |

|                  |                                                 |
|------------------|-------------------------------------------------|
| mbsf             | meters below seafloor                           |
| mbsl             | meters below sea-level                          |
| MORB             | Mid-Ocean Ridge Basalt                          |
| Mt               | Mega tons ( $\times 10^6$ )                     |
| Mtpa             | Millions of tonne per annum ( $\times 10^6$ )   |
| Myr              | Millions of year                                |
| NIST             | National Institute of Standards and Technology  |
| NOAA             | National Oceanic and Atmospheric Administration |
| ODP              | Ocean Drilling Program                          |
| OPEX             | annual OPERational EXpenditure                  |
| P                | Pressure                                        |
| pCO <sub>2</sub> | partial pressure of carbon dioxide              |
| ppm              | parts per million                               |
| PSM              | Parsons and Sclater's Model                     |
| PTS              | Polished Thin Section                           |
| RCB              | Rotary Core Barrel                              |
| RTJ              | RodriguezTriple Junction                        |
| S                | Salinity                                        |
| SCS              | Single-Channel Seismic                          |
| SEM              | Scanning Electron Microscope                    |
| SEM-EDS          | SEM with x-ray spectroscopy                     |
| STL              | Submerged Turret Loading                        |
| SWIR             | South-West Indian Ridge                         |
| T                | Temperature                                     |
| TA               | Total Alkalinity                                |
| TLT              | Temperature Logging Tool                        |
| XRD              | X-Ray Diffraction                               |
| XRF              | X-Ray Fluorescence                              |
| yr               | year                                            |
| ZEP              | Zero Emissions Platform                         |
| $\Delta\rho$     | difference of density                           |
| $\rho$           | density                                         |

# Chapter 1: Introduction

## 1.1 Anthropogenic CO<sub>2</sub> emissions into the atmosphere

*"Warming of the climate system is unequivocal" [IPCC, 2013], and "most of the observed increase in global average temperatures since the mid-20th century is extremely likely due to the observed increase in anthropogenic greenhouse gas concentrations" [IPCC, 2014b].*

Carbon dioxide (CO<sub>2</sub>) is the primary greenhouse gas emitted through human activities. But, how can we distinguish the anthropogenic sources of CO<sub>2</sub>, essentially due to the burning of fossil fuels, from the natural components? The key factor is the isotopic signature of the carbon (C) in CO<sub>2</sub>. Among the three naturally occurring isotopes of carbon on Earth (<sup>12</sup>C, <sup>13</sup>C, and <sup>14</sup>C), carbon-12 and -13 are stable. Their isotopic ratio <sup>13</sup>C:<sup>12</sup>C (usually expressed as  $\delta^{13}\text{C}$  relative to a reference material) can be used to distinguish atmospheric carbon sources. For example, plants have less <sup>13</sup>C relative to the seawater because during photosynthesis plants more easily take in <sup>12</sup>C over the heavier <sup>13</sup>C isotope. Carbon-14 (<sup>14</sup>C or radiocarbon) is a radioactive carbon isotope, with a half-life of about 5730 years. Its decay permits reliable age determination of samples up to 50k years old because after ~10 half-lives the quantity of <sup>14</sup>C remaining is <0.1% and difficult to detect. Hence, younger natural sources of CO<sub>2</sub> are relatively rich in <sup>14</sup>C and <sup>13</sup>C, whereas fossil fuels, which are usually millions of years old, are depleted in <sup>13</sup>C and contain little to no traces of <sup>14</sup>C. Considering this difference, the origin of CO<sub>2</sub> concentrations can be differentiated, making mapping of the global carbon budget possible.

The global carbon budget is the balance of the exchanges (inputs and outputs) of carbon between carbon reservoirs (carbon cycle) or within one specific loop (such as atmosphere – biosphere exchange). The Earth's oceans, soils, plants, animals and

volcanoes are all natural sources of carbon dioxide emissions, whereas human CO<sub>2</sub> inputs into the atmosphere include fossil fuel combustion (i.e. power plants, transport, and gas flaring), cement production, and land use (i.e. deforestation, logging and intensive cultivation of cropland soils). Natural and anthropogenic exchanges of carbon dioxide (CO<sub>2</sub>) between oceans, atmosphere and land are shown in Figure 1.1, as average values calculated between 2004 and 2015 [Canadell et al., 2007; IPCC, 2007; Le Quere et al., 2009; Le Quéré et al., 2014; Le Quéré et al., 2015; Sabine et al., 2004]. The total amount of CO<sub>2</sub> currently stored into the atmosphere is approximately 3100 Gt (~400 ppm). According to <sup>14</sup>C measurements, 2200 Gt (~280 ppm) are from natural sources, whereas 900 Gt (~115 ppm) are derived from human activities [www.esrl.noaa.gov, Hofmann et al., 2009]. At present, global CO<sub>2</sub> anthropogenic emissions are estimated to be 36 Gt/yr [www.cdiac.ornl.gov]. Of this, 30% is absorbed by terrestrial plants and soils ("land sink"), 26% by the oceans ("ocean sink"), with 44% remaining in the atmosphere. This leads to an annual growth of 16 Gt of CO<sub>2</sub> in the atmosphere.

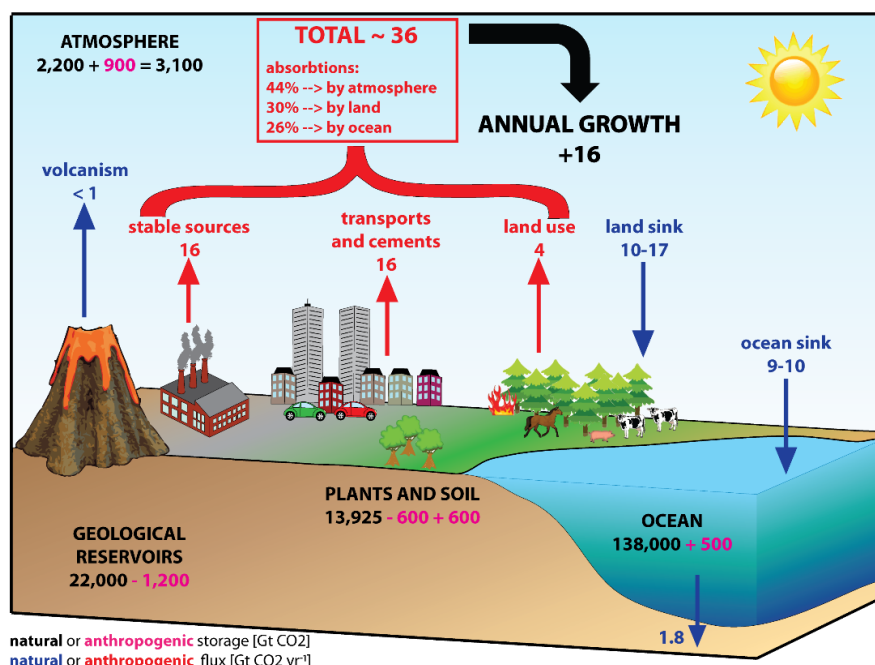


Figure 1.1: Global carbon budget shown as average values between 2004 and 2015 [Canadell et al., 2007; IPCC, 2007; Le Quere et al., 2009; Le Quéré et al., 2014; Le Quéré et al., 2015; Sabine et al., 2004]. CO<sub>2</sub> fluxes are expressed in Gt/yr with red (anthropogenic flux) and blue (natural flux) arrows. The sum of these exchanges leads to an annual growth of 16 Gt of CO<sub>2</sub> to the atmosphere. Natural and anthropogenic reservoirs of Gt of CO<sub>2</sub> in the Earth are shown in black and pink, respectively.

However, the atmospheric CO<sub>2</sub> annual growth is an average value in constant evolution. The current high concentration of CO<sub>2</sub> in the atmosphere is the result of an exponential increasing trend registered worldwide over the last 200 years (Figure 1.2). Records of atmospheric CO<sub>2</sub> concentrations measured in air bubbles trapped in ice cores [Barnola *et al.*, 1987; Luthi *et al.*, 2008] over the last 800 thousand years show a cyclic trend, corresponding to glacial and interglacial periods, with peaks and troughs always below the threshold of 300 ppm (Figure 1.2). Since the beginning of industrialisation in ~1800 AD, atmospheric CO<sub>2</sub> concentrations have risen rapidly. High CO<sub>2</sub> concentrations (>300 ppm) are only present in significantly older records. For example, during the Early Carboniferous (320-360 Ma) the atmospheric concentration of carbon dioxide was ~1000 ppm, whereas during the Late Carboniferous (270-320 Ma) CO<sub>2</sub> levels were “as low” as they are today [Franks *et al.*, 2014].

To summarize: several CO<sub>2</sub> concentration excursions during the history of our planet have been documented and dated with several geochemical proxies but no peaks above 300 ppm have occurred in the last 800 thousand years. The CO<sub>2</sub> mass in the atmosphere has constantly grown over the last 200 years, leading to a current annual increase of 16 Gt. The starting point and the development of this exponentially increasing trend coincides with the industrial use of fossil fuels on a global scale.

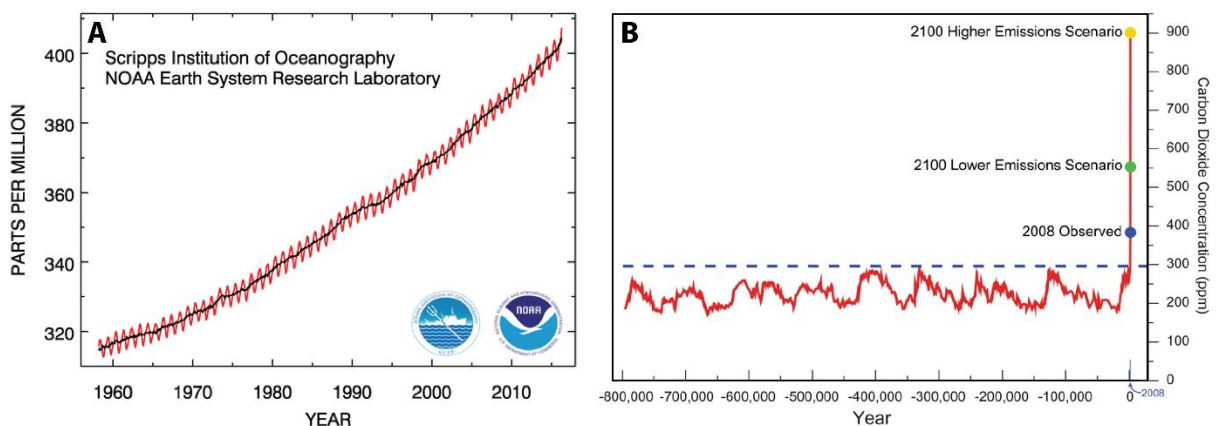


Figure 1.2: Trends of carbon dioxide concentration into the atmosphere (1 ppm of CO<sub>2</sub> = 7.84 Gt of CO<sub>2</sub>), in the last 200 years (A – “Keeling Curve” Mauna Loa Observatory May 2016, [www.esrl.noaa.gov/gmd/ccgg/trends/full.html](http://www.esrl.noaa.gov/gmd/ccgg/trends/full.html)), and over the last 800 thousands of years (B – [www.ncdc.noaa.gov](http://www.ncdc.noaa.gov)), using data from ice core records and more recent direct measurements.

## 1.2 Mitigation strategies

Climate change mitigation has the characteristics of a collective action problem at the global scale, with the goal of reducing the carbon dioxide content of the atmosphere. In the past 20 years, scientists, politicians, and economists have suggested mitigation strategies and new “green” policies to minimise the effect of this greenhouse gas on global climate and the environment (e.g. ocean acidification) [Bryant, 1997; IPCC, 2014a]. Repeated warnings sent by the United Nations Intergovernmental Panel on Climate Change, specifically in the report *IPCC* [2007], concluded that yearly reductions of at least 50% in global CO<sub>2</sub> emissions compared to 2000 levels will need to be achieved by 2050 (from ~30 to 14 Gt CO<sub>2</sub>) to limit the long-term global average temperature rise to between 2.0 and 2.4 °C relative to pre-industrial levels. The temperature limit is based on models that demonstrate how ocean circulation, weather patterns, and marine ecosystem cycles, will be irreversibly perturbed beyond this threshold [Frölicher and Joos, 2010; Prentice et al., 2001; Solomon et al., 2009], with severe implications for humans and the environment. At the Paris climate conference in 2015 (COP21), governments agreed to endeavour to limit the increase to 1.5 °C [Conference of the Parties Twenty-first Session, 2015]. On the other hand, Allen et al. [2009] and IPCC [2013] have noted that cumulative anthropogenic CO<sub>2</sub> emissions may provide a better constraint on temperature rise than CO<sub>2</sub> emission rates. Anthropogenic CO<sub>2</sub> emissions are expected to peak at 3.67 trillion tons of CO<sub>2</sub> (~470 ppm), about half of which has already been emitted, and will most likely induce a warming of 2 °C above pre-industrial temperatures, with a 5–95% confidence interval of 1.3–3.9 °C.

Various approaches have been proposed to reduce anthropogenic CO<sub>2</sub> emissions, including reducing energy demand, improving technology efficiency, and increasing the contribution of nuclear and low carbon renewable energy. An additional approach is to increase the carbon sinks, for example through Carbon Capture and Storage (CCS) of CO<sub>2</sub> in reservoirs that safely preclude the re-emission of anthropogenic CO<sub>2</sub> into the



atmosphere [Hoffert et al., 2002; IPCC, 2005b; Pacala and Socolow, 2004]. The implementation of one strategy does not preclude the deployment of others. In order to halve the current anthropogenic CO<sub>2</sub> emissions (~36 Gt/yr) and reach the target of 14 Gt/yr of CO<sub>2</sub> into the atmosphere by 2050, as indicated by the *IPCC reports*, we should apply all the possible low-carbon technologies (Figure 1.3) [IEA, 2012]. CCs is one of the most promising strategies because it could be implemented while fossil fuels remain the dominant source of energy, and renewable energies are being developed. Looking at the impact of these strategies, CCS could potentially contribute to a ~19% reduction in total CO<sub>2</sub> emissions over the next 40 years, based on the current direct emissions from industry (e.g. iron and steel, cement, chemical and petrochemical). CCS generally requires the separation of CO<sub>2</sub> from an industrial gas stream or directly from the atmosphere (Capture), compression and transport via pipelines or ships (Transport), and injection of fluids into underground geologic reservoirs (Storage).

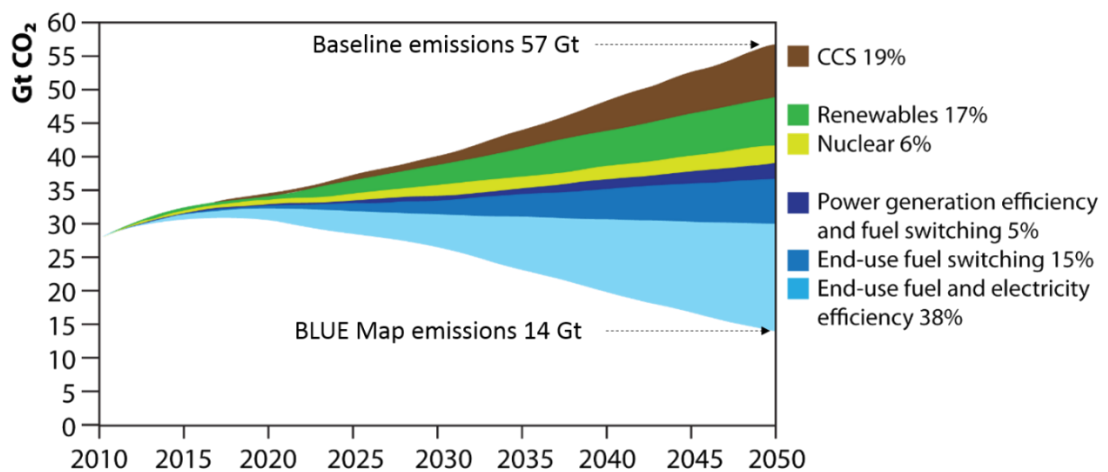


Figure 1.3: Key mitigation strategies for reducing atmospheric CO<sub>2</sub> emissions from 57 Gt (Baseline emissions) to 14 Gt (BLUE Map emissions), over the next 40 years [modified version of IEA, 2010].

### 1.3 Geological storage of CO<sub>2</sub>

Pore space in deep geological rock formations may provide a secure location to store anthropogenic CO<sub>2</sub> emissions. Several geological formations/reservoirs, such as deep

saline aquifers, depleted oil and gas reservoirs, and unmineable coal seams are being investigated as possible CO<sub>2</sub> storage reservoirs [Bachu *et al.*, 2007; IPCC, 2005a]. However, the effectiveness of these reservoirs depends on their storage capacity, reservoir stability, risk of leakage, and retention time [Benson and Cole, 2008; Hawkins, 2004; Rochelle *et al.*, 2004].

The injection of CO<sub>2</sub> into deep saline sedimentary aquifers [Eccles and Pratson, 2012; House *et al.*, 2006; Levine *et al.*, 2007; Schrag, 2009] is a particularly promising approach due to their large storage capacity, the natural presence of low permeability cap rocks, and the common proximity to major industrial sources. For example, at the Sleipner Project (240 km West of Stavanger, Norway) more than 15 million tons of CO<sub>2</sub> have been injected into a subsea saline formation in the North Sea since 1996 [[www.statoil.com/en/TechnologyInnovation/NewEnergy/Co2CaptureStorage/Pages/SleipnerVest.aspx](http://www.statoil.com/en/TechnologyInnovation/NewEnergy/Co2CaptureStorage/Pages/SleipnerVest.aspx)].

Depleted oil and gas reservoirs have also been proposed due to their large storage capacity, existing infrastructure, and because injection of gases such as carbon dioxide can be used for enhanced oil recovery [Bachu, 2000; Jessen *et al.*, 2005]. The IEA GHG Weyburn-Midale CO<sub>2</sub> Monitoring and Storage Project (Canada) is an example of CCS applied to enhanced oil recovery (EOR). This site is currently the world's largest carbon capture and storage project with over 25 million tons of injected anthropogenic CO<sub>2</sub> since the project start in 2000, and with about 2.8 million tons of CO<sub>2</sub> being stored annually [[www.ptrc.ca/projects/weyburn-midale](http://www.ptrc.ca/projects/weyburn-midale)].

The IPCC report [2005] estimates that deep saline aquifers have a storage capacity of at least 1000 Gt CO<sub>2</sub> and depleted oil and gas reservoirs could store 675–900 Gt CO<sub>2</sub>, although the storage capacity of unminable coal formations is uncertain, with estimates up to 200 Gt CO<sub>2</sub>. Considering that ~1840 Gt of CO<sub>2</sub> have already been emitted into the atmosphere (see Section 1.2.), the potential reservoir storage capacity of “conventional” formations is unlikely to accommodate future CO<sub>2</sub> anthropogenic emissions.

Hence, mafic and ultramafic formations are starting to be considered as potential sites suitable for carbon sequestration [Dessert *et al.*, 2003; Gislason and Oelkers, 2014; Gislason *et al.*, 2010; Godard *et al.*, 2011; Kelemen and Matter, 2008; Oelkers *et al.*, 2008; Power *et al.*, 2013]. This research is still at the experimental stage, and for this reason these formations are often described as “*unconventional reservoirs*” [Matter and Kelemen, 2009].

#### **1.4 Unconventional reservoirs**

Mafic and ultramafic rocks are igneous rocks rich in magnesium and iron, with 40-90% and >90% of mafic minerals (such as olivine, and pyroxene), respectively.

The importance of these rocks in carbon sequestration strategies is linked with their high content of divalent cations ( $\text{Ca}^{2+}$ ,  $\text{Mg}^{2+}$ , and  $\text{Fe}^{2+}$ ). These positively charged ions, once in contact with  $\text{CO}_2$ -rich fluids, can react with the bicarbonate ( $\text{HCO}_3^-$ ) and carbonate ( $\text{CO}_3^{2-}$ ) ions in solution, and precipitate as carbonate minerals (“mineral carbonation”), providing the permanent fixation of  $\text{CO}_2$  as a stable solid phase [e.g. Lackner *et al.*, 1995; Seifritz, 1990; Sipilä *et al.*, 2008].

Mineral carbonation can be done *ex situ* or *in situ* [IPCC, 2005b; Oelkers *et al.*, 2008].

The *ex situ* scenario is part of an industrial process where suitable initial material has to be transported to a “carbonation reactor”, ground, and heated up to precipitate the solid  $\text{CO}_2$ -rich end product. This methodology of carbonate phase formation in silicate rocks is thermodynamically favourable, but encounters numerous challenges in terms of transport, efficiency, and economic feasibility [Gerdemann *et al.*, 2007]. An example is the multi-step mineral carbonation that involves the extraction of divalent metal oxides from silicates (e.g. serpentine minerals) using industrial waste residues, and the production of carbonates using  $\text{CO}_2$  captured from stable sources located in close proximity [Dri *et al.*, 2014].

The *in situ* option overcomes the need of rock transport, by injecting a CO<sub>2</sub>-rich solution directly into porous rocks. This requires critical choices on the type of host rock formation, and the reservoir location, which should be fairly accessible and should have abundant water availability for carbonation [Kelemen and Matter, 2008].

Several studies have confirmed the natural occurrence of mineral carbonation in mafic and ultramafic rocks through the observation of carbonate veins as products of hydrothermal alteration processes in the seafloor [Alt and Teagle, 1999; Coggon *et al.*, 2010; Shibuya *et al.*, 2013] and/or surface chemical weathering [Brady, 1991]. Hence, the efficiency of mineral carbonation is the key difference between CO<sub>2</sub> storage in unconventional reservoirs versus conventional reservoirs in sedimentary basins. As shown by Dessert *et al.* [2003], natural carbonation of continental basalt consumes ~0.18 Gt of CO<sub>2</sub> per year globally. Also, multiple experiments and numerical modelling studies on CO<sub>2</sub> precipitation rates within (ultra) mafic rocks suggest that mineral carbonation could happen in the order of years [Gysi and Stefánsson, 2008; McGrail *et al.*, 2006; Paukert *et al.*, 2012; Rosenbauer *et al.*, 2012; Schaef and McGrail, 2009; Schaef *et al.*, 2010]. In sedimentary systems, however, the mineralization process would likely take from hundreds to thousands of years, due to the lack of reactive Ca, Mg, and Fe-rich silicate minerals [Benson and Cole, 2008; Gunter *et al.*, 1997], allowing the gas to potentially escape back into the atmosphere due to earthquakes or incomplete seals.

In the last decade, carbon dioxide injections into mafic and ultramafic formations have started to be considered as a valid unconventional alternative to the industrial CO<sub>2</sub> sequestration into more conventional sedimentary basins. Most of the sites recognised so far as unconventional reservoirs are located on land and characterised as flood basalts [Goldberg *et al.*, 2010; Matter *et al.*, 2007; McGrail *et al.*, 2006; Schaef *et al.*, 2010; 2011; Van Pham *et al.*, 2012], basaltic glasses [Galeczka *et al.*, 2014; Gislason *et al.*, 2010; Oelkers and Gislason, 2001], exposed peridotites [Kelemen and Matter, 2008; Kelemen *et al.*, 2011; Olsson *et al.*, 2014], or mine tailings [Harrison *et al.*, 2012; Power *et al.*, 2013; Wilson *et al.*, 2006b]. However, only two field-scale pilot CO<sub>2</sub> injection

projects have been conducted in basalts: the CarbFix Project in Iceland and the Big Sky Carbon Sequestration Partnership (BSCSP) project in U.S. [Gislason and Oelkers, 2014; Tollefson, 2013]. The CarbFix Project at the Hellisheidi geothermal power plant in SE Iceland started in 2007, and is an attempt to combine the capture of CO<sub>2</sub> from a geothermal power plant with its storage in a nearby basaltic formation ([www.carbfix.com](http://www.carbfix.com)). The gaseous CO<sub>2</sub> is dissolved in water before being injected into the subsurface formation [Alfredsson *et al.*, 2008; Gislason *et al.*, 2010; Matter *et al.*, 2009; Matter *et al.*, 2011; Oelkers *et al.*, 2008]. The pilot test involved the injection of 175 tons of pure CO<sub>2</sub>. Subsequent monitoring showed that, 95% of the injected CO<sub>2</sub> has been mineralized to carbonate minerals within less than 2 years [Matter *et al.*, 2015; Matter *et al.*, 2013; Matter *et al.*, 2016].

In the BSCSP project [McGrail *et al.*, 2006], near Wallula – Washington State, 1000 tons of supercritical CO<sub>2</sub> was injected into continental flood basalts at more than 800 m depth ([www.bigskyco2.org](http://www.bigskyco2.org)). In contrast to the CarbFix Project, in the BSCSP project the CO<sub>2</sub> is slightly less dense than reservoir water, making the presence of impermeable rock layers above the injection zone essential to keep the CO<sub>2</sub> trapped, and allow time for mineralization to occur. The first results from the fluid samples collected at the injection zone show elevated concentrations of elements such as calcium and magnesium that indicate that the injected CO<sub>2</sub> has reacted with the host rocks [McGrail *et al.*, 2014].

Only a few studies have focused on deep-sea basalts [Goldberg and Slagle, 2009; Goldberg *et al.*, 2008; Goldberg *et al.*, 2013; Slagle and Goldberg, 2011; Takahashi *et al.*, 2000]. These studies have evaluated the theoretical storage capacity based on physical properties of mid-ocean ridge flanks, and identify several potential locations for *in situ* CO<sub>2</sub> injections. However, they do not consider the phase stability of CO<sub>2</sub> and its consequences for the trapping mechanisms. No field-scale offshore pilot projects have yet been developed. Such investigations on offshore carbon sequestration are still at an early stage, leaving unsolved a series of questions:

- What are the thermodynamic conditions at the sediment-basement interface of mid-ocean ridge flanks?
- Can CO<sub>2</sub> sequestration be applied to the whole seafloor? What are the main constraints?
- Are the sites highlighted by previous studies the best locations in the oceanic crust for CO<sub>2</sub> geological storage? If so, what criteria is this based on (i.e. CO<sub>2</sub> stability over time, permeability, reduction in risk of leakages, seawater circulation...)?
- What is the effective geochemical reactivity of deep-sea basalts in a CO<sub>2</sub>-rich system? And, are the experimental results available in literature representative?
- Is CCS applied to the oceanic crust a feasible or economically advantageous strategy?
- Considering the withdrawal of the £1 billion capital subsidy by the UK government (November 2015) for two CCS projects, Peterhead and White Rose [<http://www.globalccs.institute.com>], how could CCS in deep-sea basalts be made more attractive on an industrial scale?

## 1.5 Deep-sea basalts

The oceanic crust, formed of erupted basaltic rocks placed on top of basaltic dikes and gabbro (mafic rocks), overlies peridotite (ultramafic rock) of the upper mantle. Both rock types can be found on land or offshore, with basalts covering approximately 60 % of the Earth's surface.

Given the above mentioned open questions concerning storing CO<sub>2</sub> offshore in the ocean crust (Section 1.4), this study investigates the geological storage of CO<sub>2</sub> in marine volcanic formations, specifically in deep-sea basalts that form the uppermost lavas of the oceanic crust. Despite the challenges of working in the offshore environment (such as

high costs, and difficulties in monitoring), these formations may have several advantages over other potential geological storage options. These include:

- a) large reservoir capacities, since they form most of the seafloor;
- b) *in situ* availability of seawater for gradual dissolution of CO<sub>2</sub> in porewaters into the basement [Brady and Gíslason, 1997] and for dissolved CO<sub>2</sub> injection [Kelemen et al., 2011], following the requirement that CO<sub>2</sub> must be dissolved into an aqueous solution before carbonation can begin [Gislason et al., 2014; Gunter et al., 1993; Sigfusson et al., 2015];
- c) potentially fast mineral carbonation [Elderfield et al., 1999], which in turn could guarantee fluid retention times greater than 500 years, reducing the time required for post-injection monitoring, and the possibility of accidental release [Goldberg et al., 2008; Lackner, 2003; Oelkers and Cole, 2008];
- d) low permeability sediment blankets are naturally available in some regions.

According to these parameters and the outcomes from experiments and numerical modelling on basalts, these rocks offer many of the necessary pre-requisites of extent, reactivity, and storage capacity for long-term CO<sub>2</sub> storage [Gislason and Oelkers, 2014; Goldberg et al., 2010]. However, the important question remains: how much of this storage potential is practical to use [Snæbjörnsdóttir et al., 2014]?

## **1.6 Chapter summaries**

The objectives of this thesis are to constrain the physical parameters required for trapping CO<sub>2</sub> in deep-sea basalts in order to identify possible storage targets, and to improve the fundamental understanding of geochemical reactions involved in the offshore mineral carbonation of Mid-Ocean Ridge Basalts (MORB). A combination of analytical and experimental methods are used to investigate the exploitable potential of deep-sea basalts as a CO<sub>2</sub> sequestration media.

**Chapter 2:** The thermodynamic properties at the sediment-basement interface for the whole oceanic crust are investigated and related to the phase stability of carbon dioxide at this boundary. Two case studies in oceanic crust (Juan de Fuca and eastern equatorial Pacific Ocean) are described to compare model estimations with *in situ* measurements.

**Chapter 3:** The feasibility of carbon sequestration in deep-sea basalts is explored based on physical and gravitational properties at the sediment-basement interface. Suitable targets for offshore CO<sub>2</sub> injections are evaluated, together with sites that, contrary to previous suggestions, are gravitationally unsuitable.

A summary of the results described in Chapters 2 and 3 was published as a manuscript in *Geophysical Research Letters* – Marieni et al. [2013] (Appendix CH1-1).

**Chapter 4:** The geochemical reactions associated with the dissolution of oceanic rocks during carbon storage processes are investigated through batch dissolution experiments in a CO<sub>2</sub>-seawater-rock system at CO<sub>2</sub> partial pressure of ~1 atm and 40 °C. For the first time, rocks from the upper oceanic crust and ophiolites have been used in CCS-mineral experiments.

**Chapter 5:** The costs related to the transport and storage of CO<sub>2</sub> offshore are analysed, providing several potential scenarios. Also, an attempt is made to link these cost estimations to the deep-sea basalt targets identified in the previous chapters, highlighting the technological and physical challenges.

**Chapter 6:** A general discussion on the main findings of this study and their implications in the CCS chain is carried out based on methodology, cost, and social issues. Limitations and future works are included to provide a background for further investigations of deep-sea basalt exploitation in climate change mitigation strategies.



## Chapter 2: Thermodynamic properties of CO<sub>2</sub> at oceanic crust conditions

### 2.1. Introduction

Plate tectonic theory postulates that the lithosphere, which includes the crust and the uppermost mantle, is divided into plates that move around across the Earth's surface, driven by the gravity.

The oceanic crust forms by the rifting of the oceanic lithospheric plates at the *mid-ocean ridge*. The rifting of the 100-150 km thick lithosphere causes passive upwelling of the asthenosphere and partial melting of the mantle, to produce ~7 km of oceanic crust erupted or intruded at the mid-ocean ridge. The volcanic rocks erupted onto the ocean floor are rapidly cooled by seawater circulation, producing a highly permeable, fractured upper crust. As a result, the heated and chemically changed fluids expand until they rise buoyantly, transporting significant heat and elements from the crust into the oceans by advection. This extraction of heat, known as “*hydrothermal circulation*”, significantly affects the position and geometry of magma chambers, the chemical and physical alteration of oceanic rocks, and the composition of seawater.

Detailed explanations of oceanic crust structure, heat flow estimations, and effect of hydrothermal alteration on the carbon cycle are given in the next three sub-sections.

### 2.1.1. Oceanic crust

Our current knowledge of composition and structure of the oceanic crust derives from geophysical surveys of ocean ridges [Raitt, 1963; Solomon and Toomey, 1992], ocean floor dredging and drilling [Auzende *et al.*, 1989; Becker *et al.*, 1989; Bonatti *et al.*, 1975; Cann and Funnell, 1967; Fox and Stroup, 1981; Francheteau *et al.*, 1990], and studies of *ophiolites*, relicts of uplifted oceanic crust preserved on land [Casey *et al.*, 1981; Coleman, 1977; Harper, 1984; Moores and Vine, 1971; Nicolas, 1989].

The oceanic crust is the part of Earth that is formed at mid-ocean ridges, composed of mafic rocks, and geophysically defined to be on average  $7.1 \pm 0.8$  km thick on the basis of the seismic velocity contrast with the mantle [White *et al.*, 1992]. The seismic boundary between crust and mantle is called *Mohorovičić discontinuity*, and marks a change of composition and density.

The oceanic crustal age spans from 0 to <200 Ma, with the oldest preserved in situ ocean floor found in the Western Pacific (Figure 2.1) [Müller *et al.*, 2008].

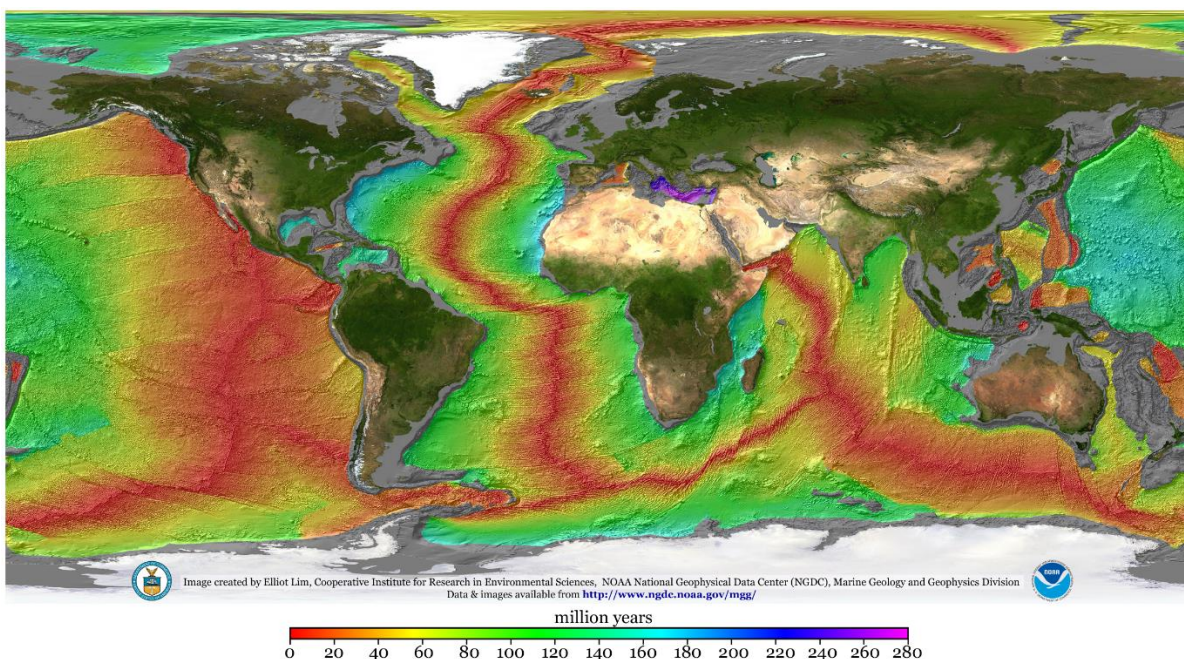


Figure 2.1: Oceanic crustal age [Müller *et al.*, 2008]. Credit for images above: CIRES & NOAA/NCEI. Scale bar extends to 280 Ma to accommodate crustal ages in the Mediterranean Sea estimated from tectonic models.

The “ideal” structure of oceanic crust is based on the Penrose ophiolite assemblage [Penrose Conference, 1972] (Figure 2.2) that includes, from top to bottom, pelagic sediments (cherts), mafic volcanic complex (pillow lavas, massive flows, and breccias), mafic sheeted dike complex, and gabbroic rocks (with several textures). All the rocks below the sedimentary layer are defined as basement, and their contact is termed the “sediment-basement interface”. In terms of physical properties, this interface separates unconsolidated rocks with low densities ( $\sim 1700 \text{ kg/m}^3$ ) [Tenzer and Gladkikh, 2014] and higher porosity ( $>30\%$ ) at the top [Hamilton and Bachman, 1982; Nafe and Drake, 1961] from crystalline rocks with higher density ( $\sim 2900 \text{ kg/m}^3$ ) and lower porosity (10%) below [Carlson and Herrick, 1990; Jarrard et al., 2003; Johnson and Pruis, 2003].

Studies of ophiolites also show that oceanic crust is underlain primarily by peridotite (Figure 2.2), called harzburgite, which is the tectonised residue left from the partial melting that produced the basalt and gabbro.

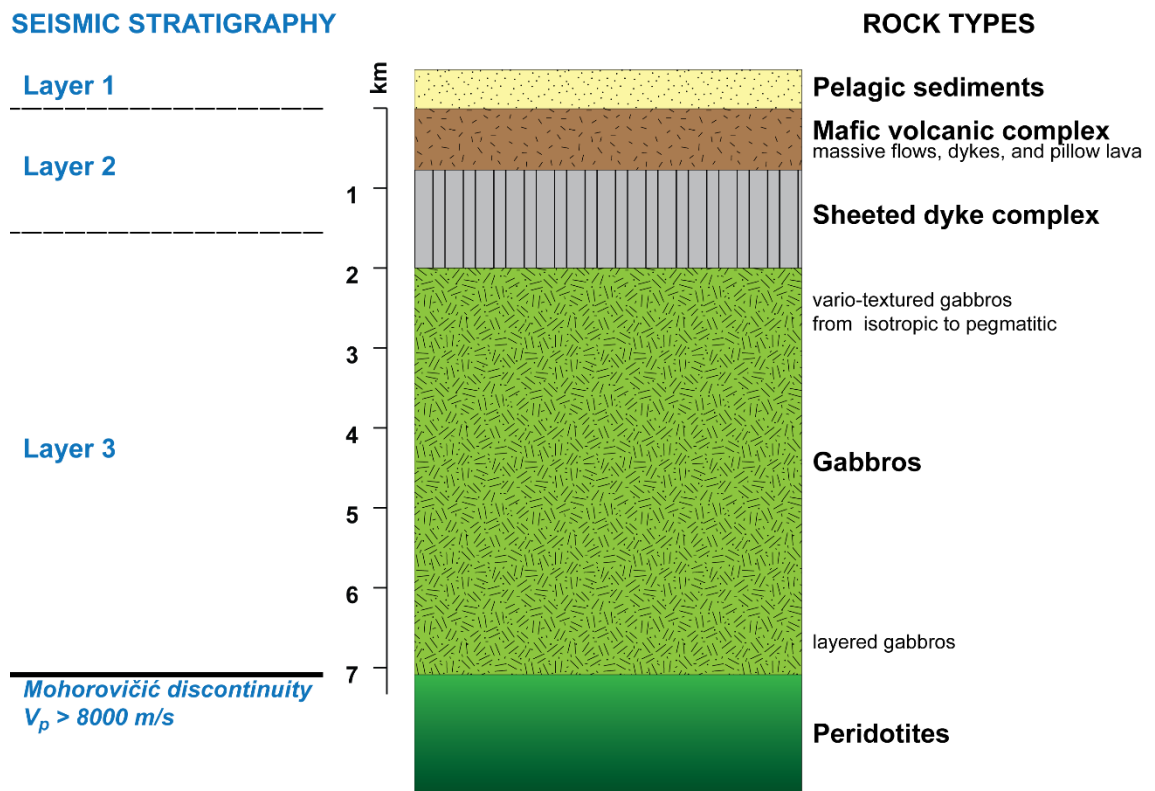


Figure 2.2: Modified version of Penrose oceanic crust stratigraphy [1972].

In general, the Penrose ophiolite model is appropriate for crust formed at fast spreading ridges, such as the East Pacific Rise, where the supply of molten material is enough to keep up with the extensional plate movements and allow compositional differentiation of lavas [Sinton and Detrick, 1992]. However, it cannot always be applied to the slow spreading oceanic crust formed at oceanic spreading rates  $<4$  cm/yr (full rate), which comprises  $\approx 50\%$  of the global mid-ocean ridge system, such as the Mid-Atlantic Ridge (MAR) and the SW Indian Ridge (SWIR) [Bach and Früh-Green, 2010]. For example, at very slow spreading ridges, the low magma supply produces thinner crust, as confirmed by seismic and gravity data, promoting the tectonic exposure of deeper rocks [Cannat, 1993; 1996]. Hence, the oceanic spreading rate is responsible for the different topography and magma supply at mid-ocean ridges.

### **2.1.2. Heat flow in oceanic crust**

Many studies have focused on the calculation of the heat budget through the oceanic crust to thermodynamically constrain the extent of conductive heat flow variation, the magnitude of the hydrothermal fluxes, and hydrothermal alteration in the basement [Alt, 1995; Lister, 1974; McKenzie, 1967; Mottl and Wheat, 1994; Parsons and Sclater, 1977; Pollack et al., 1993; Sclater and Francheteau, 1970; Sclater et al., 1980; Stein and Stein, 1992; 1994; Stein et al., 1995; Turcotte and Oxburgh, 1967; Turcotte and Schubert, 2002]. Debate on the cooling of oceanic crust started when the first heat flow measurements at sea become available in the 1960's [Langseth et al., 1966; Von Herzen and Uyeda, 1963], and 1970's [Talwani et al., 1971; Williams et al., 1974]. The main findings can be summarised as follows:

- i) New oceanic lithosphere cools, deepens, and becomes denser as it spreads away from the mid-ocean ridges, as a function of age (Figure 2.3).
- ii) The principal modes of heat transfer in the oceanic crust are conduction and advection, which can also be defined as “hydrothermal circulation”.

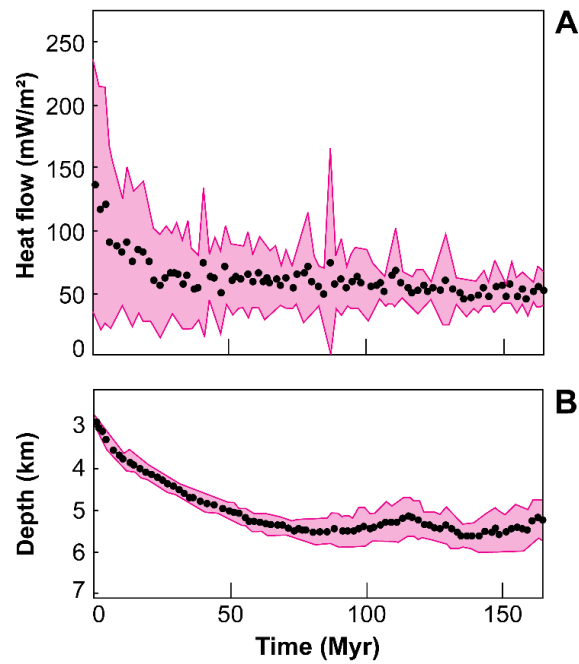


Figure 2.3: Heat flow (A) and ocean depth (B) measurements as a function of age [Stein and Stein, 1992]. Data (black dots) are averaged in 2-Ma bins, and one standard deviation about the mean value for each is shown by the contour plot (in pink).

Initially, McKenzie [1967] associated the systematic observation of the highest seafloor heat flow at mid-ocean ridges, together with the highest topographic relief, with a model explaining their exponential decrease with distance (“plate model”). Considering that the amount of cooling depends on time, Sclater and Francheteau [1970] described the data in terms of oceanic crust age rather than distance away from the ridge, providing the main constraint on the thermal structure and evolution of oceanic lithosphere. Also, Parker and Oldenburg [1973] showed that heat flow varies asymptotically as a function of  $1/\sqrt{t}$ , whereas depth increases as a function of  $\sqrt{t}$ , where  $t$  is the age of the oceanic plate.

Currently, there are three principal models to predict oceanic depth and conductive cooling in the lithosphere, and the first two are plate models with different parameters (Figure 2.4, Table 2.1):

- the Parsons and Sclater Model - PSM (in blue) [Parsons and Sclater, 1977];
- the Global Depth and Heat flow model - GDH1 (in red) [Stein and Stein, 1992];
- the Half Space Cooling Model - HSCM (in green) [Turcotte and Schubert, 1982; 2002].

Table 2.1: Summary of the key parameters used in PSM, GDH1 model, and HSCM to calculate heat flow and oceanic depth.

|                      | model   | HSCM               | PSM                  | GDH1                 |
|----------------------|---------|--------------------|----------------------|----------------------|
| Initial temperature  | °C      | 1300               | -                    | -                    |
| Thermal conductivity | W/(m K) | 3.3                | 3.1                  | 3.1                  |
| Thermal expansivity  | 1/K     | $4 \times 10^{-5}$ | $3.3 \times 10^{-5}$ | $3.1 \times 10^{-5}$ |
| Plate thickness      | km      | -                  | 125                  | 95                   |
| Basal temperature    | °C      | -                  | 1350                 | 1450                 |
| Ridge depth          | m       | 2500               | 2500                 | 2600                 |

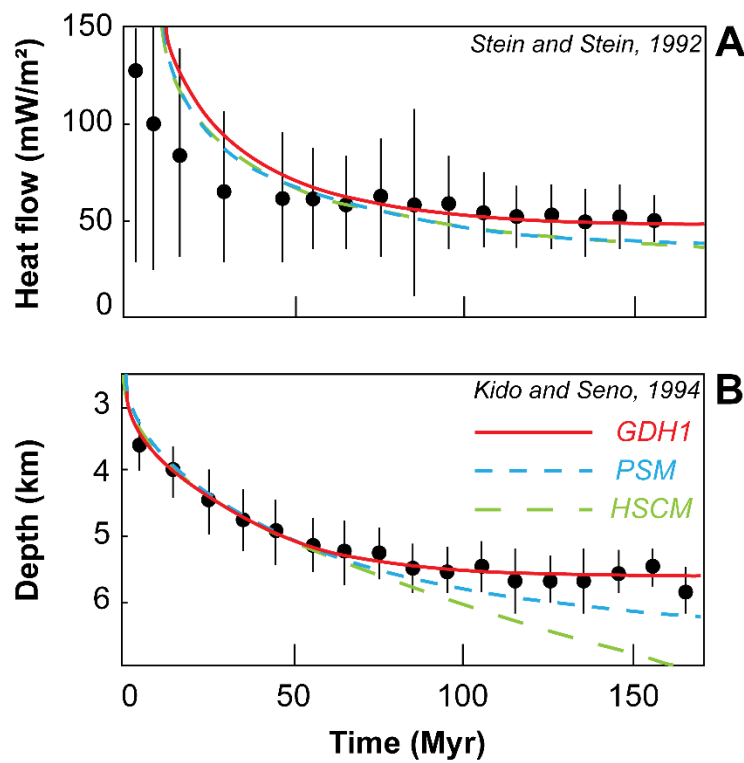


Figure 2.4: Data (black dots) and models (coloured lines) for A-heat flow and B-ocean depth as a function of age (modified from Kido and Seno [1994]; Stein and Stein [1992]; Stein et al. [1995]). The data standard deviations are represented as bars.

The PSM [Parsons and Sclater, 1977] improves on the McKenzie plate model in two main ways: data dependence on time, and slab thickness. In fact, the PSM presents the heat flow and bathymetry data as a function of the oceanic floor age (from 0 to 160 Ma) rather than distance from the ridge crest, highlighting the heat flow and depth flattening for lithospheric ages >50 Ma and >70 Ma, respectively. The PSM considers a lithosphere

where basal temperature is 1350 °C, and thickness is ~125 km – rather than the previously proposed 50 km – based on the physical parameters calculated for the deepest parts of the Pacific Ocean. However, there are numerous anomalies that were excluded from the data analysis, and hence are not explained by the model (such as seamounts, plateaus, and islands), particularly for lithosphere older than 70-100 Ma and regions with anomalously thick sediment layers.

The GDH1 model [Stein and Stein, 1992] is conceptually similar to the PSM, and addresses the misfit of *Parsons and Sclater's* model, using a larger heat flow dataset corrected for sediment thickness. As with PSM, the GDH1 model's key feature is the isothermal base of the lithosphere at a fixed depth, added to prevent the half space cooling from continuing at older ages. However, in the GDH1 model, a hotter (1450 °C at the base) and thinner lithosphere (95 km) satisfies many of the data that are anomalous in the PSM, especially for older lithosphere.

The HSCM [Turcotte and Schubert, 1982; 2002] proposes that the lithosphere is defined by a ~1300 °C isotherm within a cooling half space, which in this case is allowed to grow indefinitely. It also takes into account higher thermal coefficients of conductivity and expansivity of the crust than PSM and GDH1 models. The result is a model that shows a better agreement at younger ages (<80 Ma) than GDH1 model, but underestimates heat flow and overestimates depth for older ages. The continuing increase of depth with the age of the ocean floor defines the biggest difference with the other two models (PSM and GDH1).

In general, the GDH1 model gives a better fit on a global scale to ocean floor observations of depth and conductive heat flow than the PSM and HSCM (Figure 2.4). All models predict much higher conductive heat flow for oceanic crust <70 Ma. This discrepancy was the key evidence for the presence of heat transport by water circulation in the rocks (Figure 2.5) [Lister, 1974]. Accordingly to Stein and Stein [1994], ~34% (11

$\times 10^{12}$  W) of the total predicted global oceanic heat flux ( $32 \times 10^{12}$  W) occurs by hydrothermal flow, including both “active near ridge” and “passive off-ridge” convective regimes that extends to a crustal age of  $\sim 65$  Myr [Elderfield and Schultz, 1996]. Along the ridge axis, the temperature is  $>250$  °C, and hydrothermal circulation is vigorous [Lowell and Rona, 1985; Rosenberg et al., 1993; Strens and Cann, 1982]. On the ridge flanks, with temperature  $\ll 200$  °C, hydrothermal fluid flow is less intense, and generally restricted to the uppermost few hundred meters of permeable volcanic rocks. However, because of the vast expanse of the ocean ridge flanks, the more extended flank circulation accounts for  $\sim 70\%$  of the total hydrothermal heat loss [Becker et al., 1989; Fehn et al., 1983; Fisher et al., 1990; Mottl and Wheat, 1994].

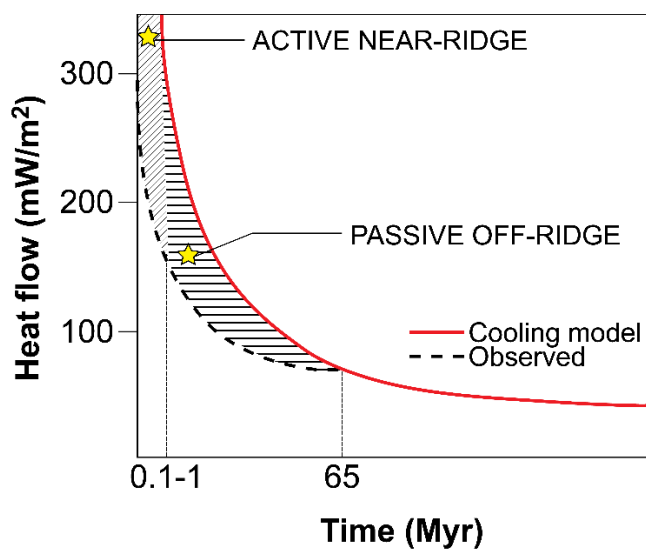


Figure 2.5: Schematic illustration of observed (dashed line) and estimated (red line) heat flow, with difference defined as a hydrothermal flux (dashed area) [modified from Stein and Stein, 1994].

The seawater flux necessary to produce the hydrothermal heat loss from the oceanic crust is  $\sim 1 \times 10^{15}$  kg/yr [Harris and Chapman, 2004], where only  $\sim 7\%$  of it is occurring at near-axial regions [Stein and Stein, 1994]. During hydrothermal alteration, seawater penetrates into the basement and reacts with volcanic rocks, resulting in major-ion composition changes in seawater dominated by: i) loss of  $\text{Mg}^{2+}$  – which is fixed in the crust as clay minerals, filling fractures and replacing volcanic glass, ii) addition of  $\text{Ca}^{2+}$  –



which is dissolved from basalts , iii) subsequent loss of alkalinity – due to the precipitation of secondary minerals, such as clays, sulphate, and carbonate minerals [Fisher *et al.*, 1990; Mottl, 1983; Mottl and Wheat, 1994].

### **2.1.3. Oceanic floor carbon uptake**

Studies on basement fluids and secondary minerals in the oceanic crust have shown that one major effect of hydrothermal circulation is the uptake of carbon during the alteration of the volcanic section [Alt and Teagle, 1999; Coggon *et al.*, 2004; Coogan and Gillis, 2013; Gillis and Coogan, 2011; Staudigel *et al.*, 1989]. A portion of the carbon dioxide (CO<sub>2</sub>) present in seawater is consumed by dissolution of oceanic crust silicates, and subsequent precipitation of carbonate minerals like calcite (CaCO<sub>3</sub>).

Mass balance models indicate that the production rate of carbon in new crust due to volcanic outgassing at mid-ocean ridges is lower than the amount of carbon fixed in the oceanic crust per year, which is  $\sim 3 \times 10^{12}$  mol C/yr (0.036 Pg C/yr) [Alt and Teagle, 1999; Staudigel *et al.*, 1989]. Consequently, the seafloor alteration is a net sink of carbon in oceanic crust ( $1.5\text{-}2.4 \times 10^{12}$  mol C/yr, or 0.018-0.029 Pg C/yr) [Alt and Teagle, 1999], and plays an important role in the current debate on the global carbon cycle [Kelemen and Manning, 2015] raising a number of questions:

- How much CO<sub>2</sub> can be removed from seawater and mineralised?
- What are the optimal conditions for CO<sub>2</sub> consumption in the basement?
- What is the CO<sub>2</sub> phase stability in the seafloor?

In this chapter, I will consider the latter question, combining information from several global databases to determine the thermodynamic properties in the oceanic crust at the sediment-basement interface and to investigate the stability of carbon dioxide at this boundary.

## 2.2. CO<sub>2</sub> and seawater

Carbon dioxide is the most important greenhouse gas after water vapour in the Earth's atmosphere, and more than 98% of the carbon dioxide in the ocean-atmosphere system is stored in the oceans as dissolved inorganic carbon [Zeebe and Wolf-Gladrow, 1993]. CO<sub>2</sub> has no liquid state at pressures below 0.52 MPa. At 0.1 MPa (~1 atm), CO<sub>2</sub> condenses directly as a solid at temperatures below -78.5 °C, and vice versa the solid undergoes sublimation (Figure 2.6). The CO<sub>2</sub> triple point, where the three phases of CO<sub>2</sub> coexist in thermodynamic equilibrium, is at 0.52 MPa and -56.6 °C. The critical point, above which there is no phase boundary between gas and liquid, is at 7.38 MPa at 31.1 °C. Above this pressure and temperature limit, carbon dioxide behaves as a supercritical fluid, which combines properties of gas and liquid (Figure 2.6). A supercritical fluid has higher diffusivity, lower viscosity, and lower surface tension than the liquid phase. This means that supercritical CO<sub>2</sub> has, in general, better solvent properties than liquid CO<sub>2</sub>.

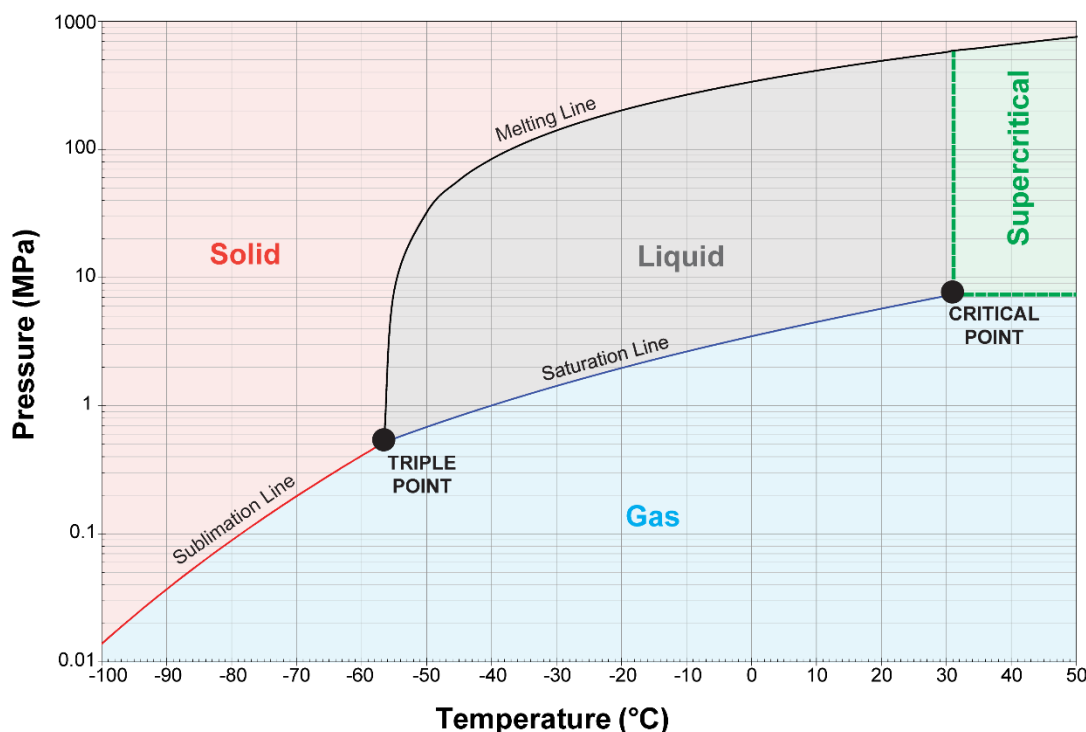


Figure 2.6: Phase diagram of CO<sub>2</sub> for temperatures from -100 to 50 °C, and pressures from 0.01 and 1000 MPa (modified from <http://www.chemicallogic.com>).

Seawater is a complex mixture of water, salts, and smaller quantities of other substances, including dissolved inorganic and organic materials and dissolved atmospheric gases. On average, seawater in the World's ocean has a salinity of ~35 practical salinity units – psu – or grams of salt dissolved in 1 kg of seawater [Brown et al., 1995]. The six ions that make up 99% of all sea salts, from the most to the least abundant, are: *chloride* ( $\text{Cl}^-$ ), *sodium* ( $\text{Na}^+$ ), *magnesium* ( $\text{Mg}^{2+}$ ), *sulfate* ( $\text{SO}_4^{2-}$ ), *calcium* ( $\text{Ca}^{2+}$ ), and *potassium* ( $\text{K}^+$ ) [Millero et al., 2008; Summerhayes and Thorpe, 1996]. Because of its salt content, seawater has different properties than fresh water, such as higher density and thermal conductivity, but lower specific heat capacity [Sharqawy et al., 2010].

For the purpose of this thesis, the parameter that best describes the differences between  $\text{CO}_2$  and seawater is density, which is directly affected by changes in temperature (T) or pressure (P). The densities of both compounds are calculated for pressures from 0 to 60 MPa and temperatures from 0 to 100 °C (Figure 2.7). The density of  $\text{CO}_2$  ( $\rho_{\text{CO}_2}$ ) is determined by interpolating the online National Institute of Standards and Technology (NIST) database [Linstrom and Mallard, 2012], which is based on the equation of state by Span and Wagner [1996]. Seawater density ( $\rho_{\text{seawater}}$ ) is estimated using the SeaWater MATLAB library [Fofonoff et al., 1983], assuming a constant salinity of 35 psu [Brown et al., 1995].  $\text{CO}_2$  density decreases dramatically with decreasing pressure and increasing temperature, compared to a near constant density for seawater (Figure 2.7). Furthermore, it is evident that carbon dioxide is denser than seawater only for pressures higher than 27 MPa (~2700 m of water) and temperature between 0 and 30 °C. Note that within this limited P-T window, liquid  $\text{CO}_2$  is the thermodynamically stable phase. Liquid  $\text{CO}_2$  has only a narrow range of densities - between 1040 and 1125  $\text{kg/m}^3$ , whereas the density of supercritical  $\text{CO}_2$  varies from 140  $\text{kg/m}^3$  (at 100 °C, 8 MPa) to 1045  $\text{kg/m}^3$  (at 30 °C, 60 MPa).

In the next Section, the phase stability of CO<sub>2</sub> and seawater at the sediment-basement interface throughout all the oceans is determined as a function of the actual temperatures and pressures. Also, the effect of heat flow and sediment thickness variations at the top of the basement, in a world-wide scenario, is evaluated.

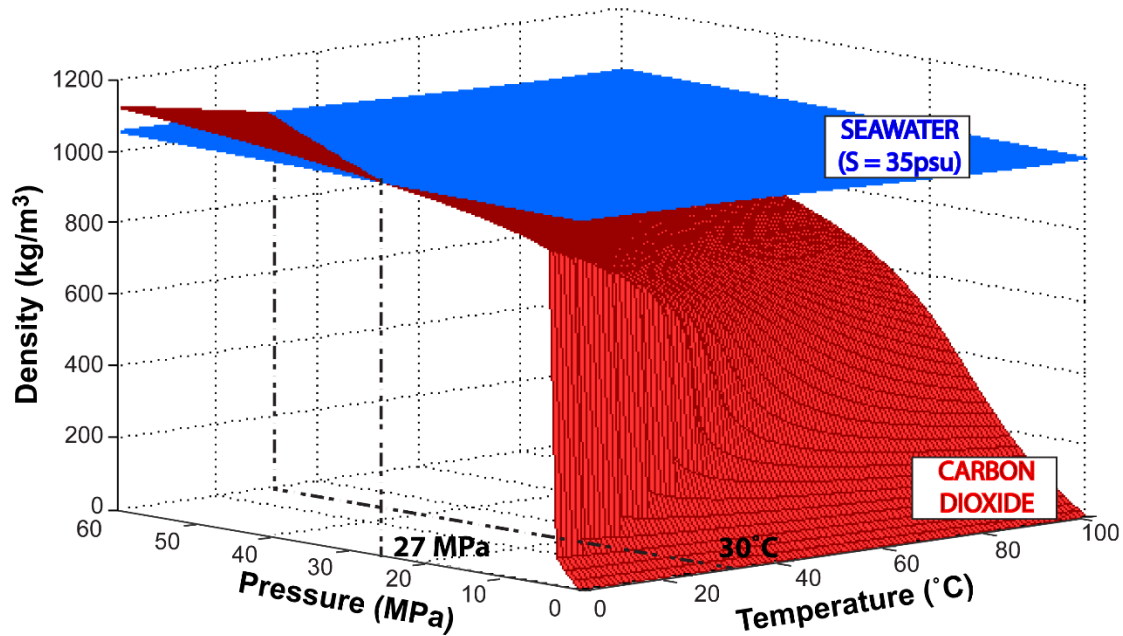


Figure 2.7: Densities of seawater (in blue) and carbon dioxide (in red) as a function of temperature between 0 and 100 °C, and pressure between 0 and 60 MPa.

## 2.3. Thermodynamic properties of oceanic crust

### 2.3.1. Heat flow estimation for the oceanic lithosphere

As previously described in Section 2.1.2, it is widely recognized that the GDH1 model provides a significantly better fit to observed conductive heat flow data, especially for older lithosphere where GDH1 gives values of heat flow higher than the estimates from the HSCM (Figure 2.4). Hence, in this study, GDH1 is considered the most appropriate model to evaluate the effect of heat flow at the sediment-basement interface throughout the world oceans. The PSM is not considered because the GDH1 model represents a development of the PSM. Conductive heat flow has been calculated for the global oceanic floor using the models GDH1 and HSCM.

The equations to calculate heat flow and bathymetry are listed for the GDH1 and HSCM models in Table 2.2. Oceanic crustal age is the only required input for the GDH1 model, whereas the HSCM also requires knowledge of lithospheric thermal conductivity (assumed to be 3.3 W/(m K)), temperature difference between the base and top of the lithosphere (assumed to be 1300 K), and thermal diffusivity (assumed to be  $10^{-6}$  m<sup>2</sup>/s).

Table 2.2: Summary of the equations used by GDH1 (left) and HSCM (right) to calculate heat flow and bathymetry.

|                   | <b>GDH1</b><br>[Stein and Stein, 1992]                                                                                                    | <b>HSCM</b><br>[Turcotte and Schubert, 1982; 2002]                                                                                                                                                                                                                                           |
|-------------------|-------------------------------------------------------------------------------------------------------------------------------------------|----------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------------|
| <b>Heat flow</b>  | $\begin{cases} t \leq 55 \text{ Ma}, & q_0 = 510/\sqrt{t} \\ t > 55 \text{ Ma}, & q_0 = q_a + 96 \exp\sqrt{-0.0278 * t} \end{cases}$      | $q_0 = K (T_1 - T_0) / \sqrt{\pi k t}$                                                                                                                                                                                                                                                       |
|                   | $q_0$ : surface heat flow [mW/m <sup>2</sup> ]<br>$t$ : oceanic crustal age [Ma]<br>$q_a$ : asymptotic heat flow [48 mW/m <sup>2</sup> ]  | $q_0$ : surface heat flow [mW/m <sup>2</sup> ]<br>$K$ : lithospheric thermal conductivity [3.3 W/(m K)]<br>$(T_1 - T_0)$ : thermal difference between base and top of the lithosphere [1300 K]<br>$k$ : thermal diffusivity [ $10^{-6}$ m <sup>2</sup> /s]<br>$t$ : oceanic crustal age [Ma] |
| <b>Bathymetry</b> | $\begin{cases} t < 20 \text{ Ma}, & dw = dr + 365\sqrt{t} \\ t \geq 20 \text{ Ma}, & dw = 5651 - 2473 \exp\sqrt{-0.0278 * t} \end{cases}$ | $dw = dr + 350 \sqrt{t}$                                                                                                                                                                                                                                                                     |
|                   | $dw$ : bathymetry [m]<br>$t$ : oceanic crustal age [Ma]<br>$dr$ : ridge depth [2600 m]                                                    | $dw$ : bathymetry [m]<br>$t$ : oceanic crustal age [Ma]<br>$dr$ : ridge depth [2500 m]                                                                                                                                                                                                       |

To determine the heat flow at the sediment-basement interface for the whole oceanic lithosphere, the oceanic crustal age database [Müller *et al.*, 2008] has been adapted for a grid with 6 minute resolution (measurement nodes every ~10 km) (Figure 2.1). The consequent age-derived heat flow values, generated from both the HSCM and GDH1 models, have been considered only if they are below a reasonable threshold of 500 mW/m<sup>2</sup> (the mean value of oceanic heat flow measurements is 101 mW/m<sup>2</sup>), and plotted on global maps, using the GMT-*Generic Mapping Tools software* (Appendices CH2-1 A and B).

### 2.3.2. Temperature at the sediment-basement interface of the oceanic lithosphere

Following *Heberling et al.* [2010], from both the heat flow databases (GDH1 and HSCM), the geothermal gradient has been calculated for a grid of 6' x 6' using Fourier's Law:

$$\frac{dT}{dz} = \frac{q_0}{K_s}$$

where  $dT/dz$  is the geothermal gradient [K/m],  $q_0$  the heat flow [W/m<sup>2</sup>], and  $K_s$  the thermal conductivity of marine sediments, taken as 1 W/(m K). To calculate the temperature at the sediment-basement interface from the geothermal gradient, which is the rate of increasing temperature with respect to depth, knowledge of sediment thickness is required. Hence, the sediment thickness for all the oceanic crust has been computed from the NOAA-gridfive dataset [*Divins*, 2003] (Appendix CH2-2), and the temperature at the top of the basement has been extrapolated from both HSCM and GDH1 model and presented as global maps (Appendices CH2-3A and B, respectively). In general, the calculation with the GDH1 model gives temperatures higher than with the HSCM due to different lithospheric isothermal constraints associated with each model (Section 2.1.2).

The use of  $K_s$  as a constant value (1 W/(m K)) is based on two major global databases of marine sediment thermal conductivities: **A)** the IODP-ODP database with the most up to date data [*International Ocean Discovery Program*, 2015] and **B)** the historical global compilation of *Pollack et al.* [1993]. These databases give the global average values of  $K_s = \sim 1.3$  and  $\sim 1$  W/(m K), respectively (Figure 2.8).

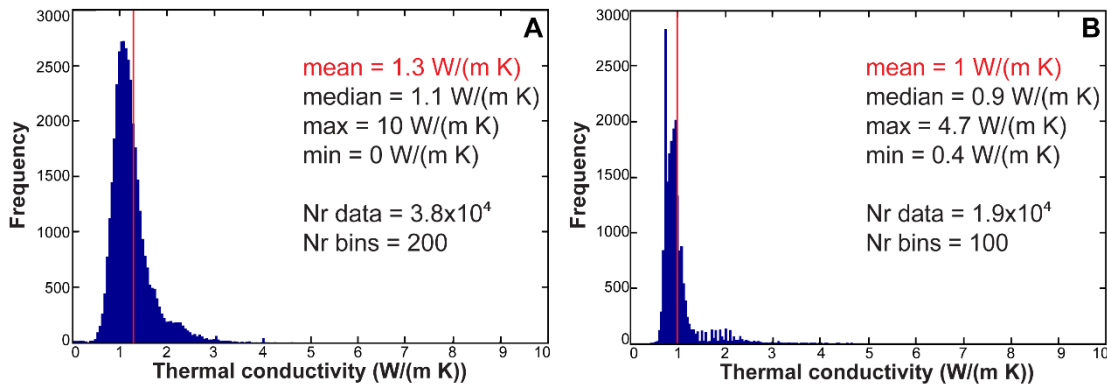


Figure 2.8: Histograms of thermal conductivity measurements for marine sediments from A) the IODP-ODP database [*International Ocean Discovery Program*, 2015], and B) the global compilation [*Pollack et al.*, 1993]. The red line shows the mean, which is close or equal to 1 W/(m K) for both databases.

### 2.3.3. Pressure at the sediment-basement interface of the oceanic lithosphere

In general, there are two approaches to estimate the pressure at the sediment-basement interface, depending on the assumption of hydrostatic or lithostatic load above the basement.

- 1) Hydrostatic, considers all the load as water column:  $u = g \cdot \rho_w \cdot (d_w + d_s)$ ;
- 2) Lithostatic, considers both overlaying rock mass and water column:

$$u = (g \cdot \rho_w \cdot d_w) + (g \cdot \rho_s \cdot d_s);$$

where  $g$  is the gravity constant of  $9.81 \text{ m/s}^2$ ,  $\rho_w$  and  $\rho_s$  the seawater and sediment density [ $\text{kg/m}^3$ ], respectively,  $d_w$  the water depth [m], and  $d_s$  the sediment thickness [m] [Mann and Mackenzie, 1990]. Based on Brown *et al.* [1995], seawater has been considered in its average conditions, with a constant density of  $1030 \text{ kg/m}^3$ , and a constant salinity of 35 psu. Measured marine sediment density values are on average  $1700 \text{ kg/m}^3$  [Tenzer and Gladkikh, 2014]. Consequently, the lithostatic pressure increases more rapidly with sediment thickness than hydrostatic pressure.

This difference between pressure assumptions based on the sediment thickness highlights the importance of the choice on the pressure assumption to use. The chosen pressure should represent the vertical stress to which the fluid (e.g.,  $\text{CO}_2$ , seawater) would be subjected to in the pore spaces, assuming an interconnected fluid network. This type of pressure is defined as effective pore pressure in Terzaghi's principle [Terzaghi *et al.*, 1996]:

$$\sigma = \sigma' + u$$

in which the total stress ( $\sigma$ ) is composed of two parameters,  $\sigma'$  the effective stress and  $u$  the effective pore pressure. More specifically, the effective stress is interpreted as the stress that the solid granules constituting the sediment exchange at the contact points. The effective pore pressure is defined as the stress acting on the water in every direction with equal intensity, and it is calculated in the same way as the hydrostatic pressure. Hence, the hydrostatic assumption will be used in all of the following calculations involving the pressure in the oceanic crust basement.

The water column thickness, taken to be the sum of the water depth and sediment thickness, has been calculated at each point on a 6-minute resolution grid by combining two databases:

- the General Bathymetric Chart of the Oceans (GEBCO) – gridfive world bathymetry map [IOC *et al.*, 2003] (Appendix CH2-4);
- the NOAA-gridfive sediment thickness database (Appendix CH2-2).

Although anomalies in seafloor topography due to seafloor abyssal hills and seamounts are not always detected by global altimetry and gravity analyses, the NOAA sediment thickness database is currently the best available. For this study, only sediment thicknesses between 0 and the reasonable threshold of 2000 m are considered. A water depth  $>0$  m is the only requirement for the bathymetry. The resulting hydrostatic pressure calculated for the whole oceanic crust at the sediment-basement interface ranges from 0 to 104 MPa, and has been presented as a global map (Appendix CH2-5).

It should be noted that the pressure estimations are completely independent from the heat flow model chosen for the thermal calculations.

#### **2.3.4. CO<sub>2</sub> density at the sediment-basement interface of the oceanic lithosphere**

Density ( $\rho$ ) is the parameter that best describes the buoyancy of CO<sub>2</sub> at the different temperature and pressure conditions calculated for the oceanic lithosphere. CO<sub>2</sub> density decreases dramatically with decreasing pressure and increasing temperature, compared to a near constant density for seawater (Figure 2.7).

Densities for CO<sub>2</sub> and seawater have been determined at the sediment-basement interface of the whole oceanic lithosphere, at each point on a 6' x 6' grid, applying the density databases described in Section 2.2 to the temperature and pressure values calculated with the GDH1 and HSCM models (using  $K_s = 1$  W/(m K) for both). The relative



density difference defined between CO<sub>2</sub> and seawater is calculated as a density difference ( $\Delta\rho$ ) on global maps for both models (Figure 2.9 and 2.10, for map generated with GDH1 and HSCM, respectively). There is little significant difference between the two models. Positive  $\Delta\rho$  ( $\rho_{CO_2} > \rho_{seawater}$ ) indicates more dense CO<sub>2</sub>. This only occurs at hydrostatic pressure higher than 27 MPa and temperatures between 0 and 30 °C. These conditions are typical of old oceanic crust in deep oceans. In contrast, negative  $\Delta\rho$  ( $\rho_{CO_2} < \rho_{seawater}$ ) describes low density CO<sub>2</sub>, which occurs at low pressure and high temperature, characteristic of shallow, young crust. The distribution of  $\Delta\rho$  on the maps is controlled by a combination of basement heat flow and sediment thickness variations:

- high heat flow rate leads to high upper crustal temperature;
- high sediment thickness corresponds to high pressure, but also to high temperature, due to the hydrologic isolation effect of rapidly sedimented pelagic deposits that with their low-permeability trap the heat at the sediment-basement interface [Davis *et al.*, 1992; Fisher and Davis, 2000; Langseth *et al.*, 1988; Stein *et al.*, 1995].

Hence, most of the areas close to a ridge axes – loci of freshly created oceanic crust – or nearby continental margins – regions of thick layers of sediments – show negative  $\Delta\rho$  conditions. However, there are some exceptions; for example, at the Rodriguez Triple Junction (RTJ) in the Indian Ocean, where the wide depression created by the cross of three rift valleys at water depth >3600 m and sediment thickness <100 m [Munsch and Schlich, 1989] causes a unique P-T environment, enabling the density of CO<sub>2</sub> to be greater than seawater at the ocean floor.

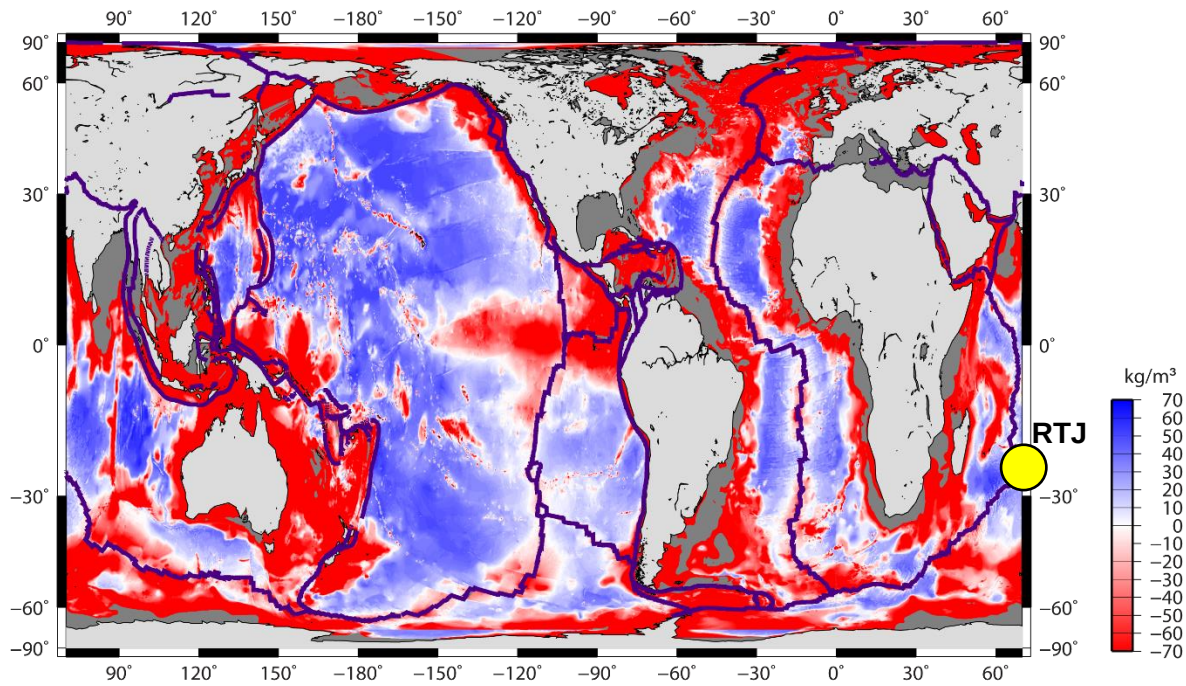


Figure 2.9: Map of density difference  $\Delta\rho$  between  $\text{CO}_2$  and seawater at the sediment-basement interface, generated using the GDH1 model, with  $K_s = 1 \text{ W/(m K)}$ . The red shadings show negative  $\Delta\rho$  ( $\rho_{\text{CO}_2} < \rho_{\text{seawater}}$ ); the blue show positive  $\Delta\rho$  ( $\rho_{\text{CO}_2} > \rho_{\text{seawater}}$ ). The RTJ-Rodriguez Triple Junction is indicated with a yellow circle.

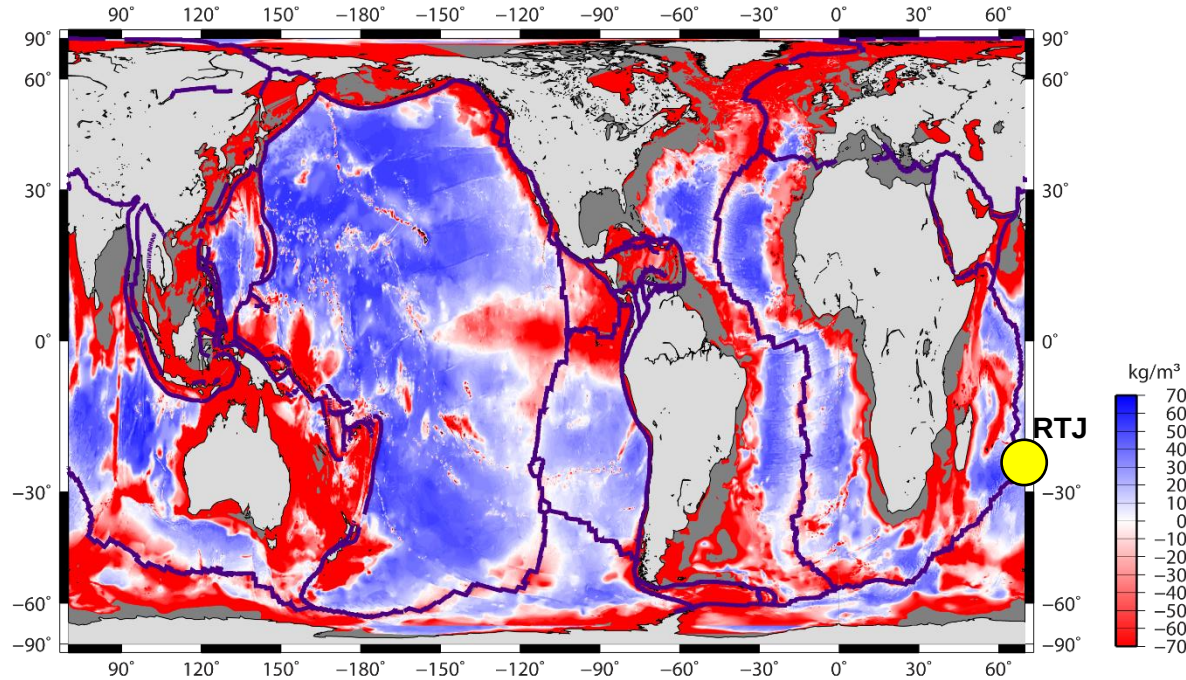


Figure 2.10: Map of density difference  $\Delta\rho$  between  $\text{CO}_2$  and seawater at the sediment-basement interface, generated using the HSCM model, with  $K_s = 1 \text{ W/(m K)}$ . The red shadings show negative  $\Delta\rho$  ( $\rho_{\text{CO}_2} < \rho_{\text{seawater}}$ ); the blue show positive  $\Delta\rho$  ( $\rho_{\text{CO}_2} > \rho_{\text{seawater}}$ ). The RTJ-Rodriguez Triple Junction is indicated with a yellow circle.

## **2.4. Thermodynamic properties of two case studies in oceanic crust**

The previously described findings on T, P, and CO<sub>2</sub> densities at the sediment-basement interface in a world-wide scenario are the result of the application of heat flow models and the interpolation of several databases. To validate their interpretations, the model estimations have to be compared with actual measured temperatures and pressures at the sediment-basement interface. To address this issue, available *in situ* measurements, laboratory analyses and model outputs have been considered for two well-studied oceanic locations: the Juan de Fuca Plate (JdFP) and the eastern equatorial Pacific Ocean (eePO). Although there are numerous holes drilled into the oceanic crust by scientific ocean drilling, JdFP and eePO are the only locations where accurate temperatures close to the sediment-basement interface have been measured.

### **2.4.1. Juan de Fuca Plate**

The Juan de Fuca Plate (JdFP – Figure 2.11), offshore Washington State and British Columbia (USA), has been the focus of conceptual studies of Carbon Capture and Storage (CCS) applied to deep sea basalt [Goldberg *et al.*, 2008] because it is one of the best studied mid-ocean ridge flanks, and it is near to major industrial regions of North America. The JdFP has been the locus of three scientific ocean drilling cruises (ODP Leg 168, and IODP Expeditions 301 and 327) and numerous other scientific experiments that have defined the regional thermal and hydrological regimes of the upper oceanic crust [Becker *et al.*, 2013; Coggon *et al.*, 2004; Davis *et al.*, 1997; Fisher and Davis, 2000; Wheat and Mottl, 2000; Wheat *et al.*, 2003].

The oceanic crust on this plate is relatively young, having formed at the Juan de Fuca Ridge between 0 and 11 million years-ago at intermediate spreading rate (5-7 cm/yr full rate) [Govers and Meijer, 2001], and occurs over a water depth range of 2200 to 3500 m. The pillow lavas that form the upper few hundred meters of the JdFP crust have high connected porosity (>10%), which allows to cold, unaltered seawater to enter the crust where basement is exposed at the seafloor [Fisher, 1998].

The ridge flank is blanketed by a very thick (from ~30 to >700 m) sequence of hemipelagic and turbiditic sediments derived from the North American continental margin during the Pleistocene (2.58 – 0.01 Ma) [Fisher and Davis, 2000].

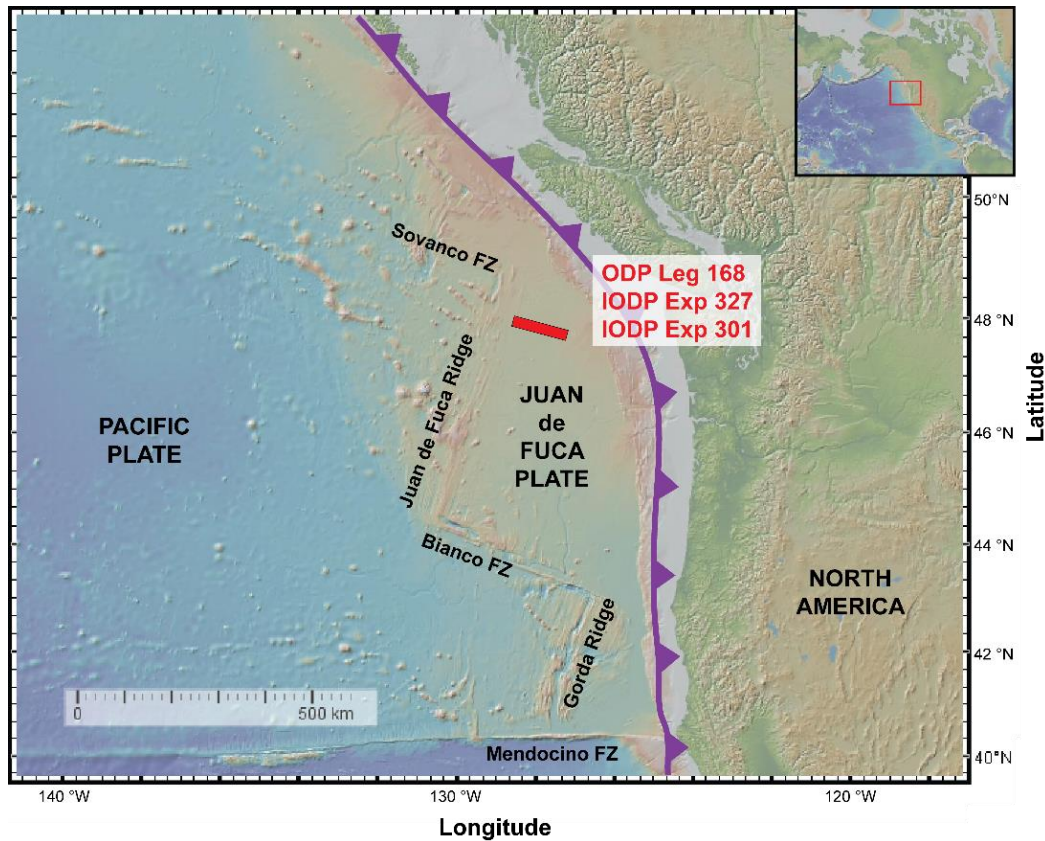


Figure 2.11: Tectonic setting of the Juan de Fuca Plate (JdFP). The sites of interest are highlighted in red.

The eastern ridge flank of the JdFP was investigated during the ODP Leg 168 (Figure 2.12). Ten sites were drilled between 20 and 100 km from the ridge axis, in crust ranging from 0.8 to 3.6 Ma, and sediment thickness increasing from 40 to 600 m (summary in Table 2.3). This relatively continuous and anomalously thick sediment cover has been found to be responsible for confining the hydrothermal fluids within the igneous basement for most of the ridge flank, and consequently affecting the thermal setting along the transect [Davis *et al.*, 1997].



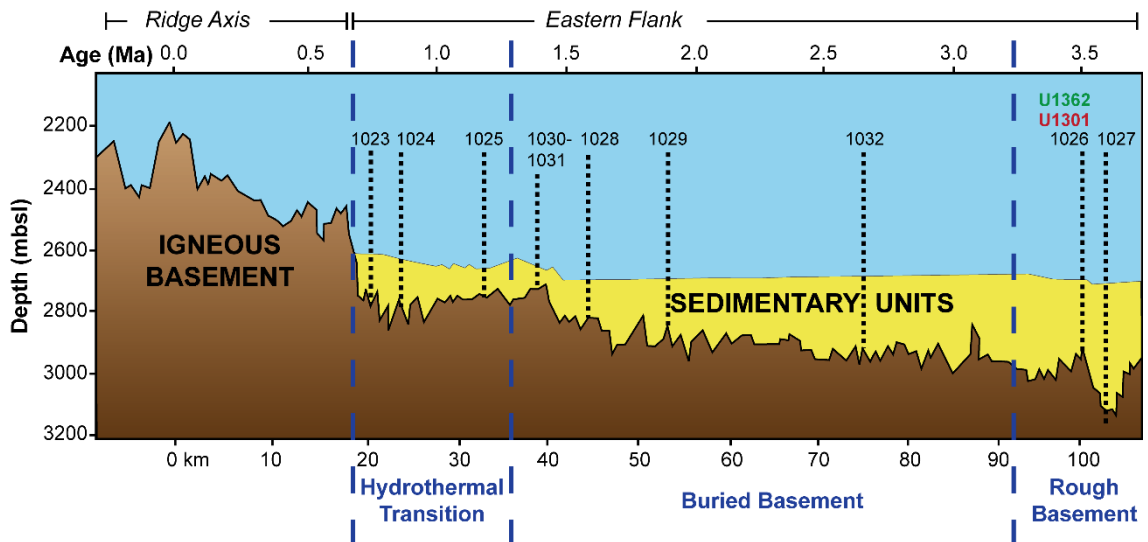


Figure 2.12: Schematic cross section of the ODP Leg 168 drilling transect profile (derived from seismic reflection), adapted from Davis et al. [1997]. Drill site locations of ODP Leg 168 (in black), IODP Exps 301 (in red) and 327 (in green), and inferred basement age are shown.

### Temperature

The temperatures of pore water and basement fluids at the sediment-basement interface increase with distance from the ridge and range from 16° to 63°C (summary in Table 2.3) [Coggon et al., 2004; Elderfield et al., 1999; Pribnow et al., 2000]. The high temperatures at the sediment-basement interface have been explained by the blanket effect of the anomalously thick sediment cover over young oceanic crust [Davis et al., 2004; Fisher et al., 2003; Stein et al., 1995].

### Pressure

As previously mentioned (Section 2.3.3), the hydrostatic pressure assumption is the most appropriate approach to describe the vertical stress to which the CO<sub>2</sub> would be subjected at the boundary between sediments and oceanic crust basement. Hence, the corresponding hydrostatic pressures of each ODP Leg 168 site at the sediment-basement interface have been determined, and compared to lithostatic estimations to highlight the pressure discrepancy.

The fundamental parameters considered in the calculations are:

- seawater density, locally calculated to be 1027 kg/m<sup>3</sup>;
- sediment density, ranging between 1800 and 2000 kg/m<sup>3</sup> [Davis *et al.*, 1997];
- water depth, which on average corresponds to ~2600 m [Davis *et al.*, 1997];
- sediment thickness, which was measured for each site during the drilling expedition.

The hydrostatic and lithostatic pressures range from 26 to 33 MPa and from 26 to 39 MPa, respectively, showing a good agreement for most of the sites (summary in Table 2.3). The maximum pressure difference of 6 MPa has been obtained for a single site (Site 1027) characterised by sediment thickness greater than 600 m, whereas the average pressure difference for all the other sites, with sediment thickness below 300 m, is 1.4 MPa.

*Table 2.3: Summary of the properties of the sites on JdFP drilled during the ODP Leg168. Pressure - P (MPa) and temperature (°C) conditions are calculated at the sediment-basement interface. Distance from ridge axis, oceanic crustal age, water depth, and sediment thickness are also shown.*

| ODP Leg 168 Sites             | 1023 | 1024 | 1025 | 1030 | 1031 | 1028 | 1029 | 1032 | 1026  | 1027  |
|-------------------------------|------|------|------|------|------|------|------|------|-------|-------|
| Distance from ridge axis (km) | 22.4 | 25.6 | 33.6 | 39.4 | 39.4 | 44.8 | 54.8 | 74.6 | 101.1 | 103.3 |
| Oceanic crustal age (Ma)      | 0.86 | 0.97 | 1.24 | 1.43 | 1.43 | 1.62 | 1.95 | 2.62 | 3.51  | 3.59  |
| Water depth (m)               | 2593 | 2614 | 2602 | 2574 | 2588 | 2659 | 2653 | 2645 | 2658  | 2657  |
| Sediment thickness (m)        | 193  | 168  | 96.6 | 41.9 | 41.3 | 133  | 220  | 290  | 229   | 614   |
| Temperature (°C)              | 15.5 | 22.8 | 38.6 | 40.1 | 40.4 | 50.5 | 58.7 | 57.1 | 61.7  | 62.8  |
| P hydrostatic (MPa)           | 28.1 | 28.0 | 27.2 | 26.3 | 26.5 | 28.1 | 28.9 | 29.6 | 29.1  | 32.9  |
| P lithostatic (MPa)           | 29.8 | 29.5 | 28.1 | 26.7 | 26.8 | 29.3 | 30.9 | 32.0 | 31.3  | 38.9  |

#### 2.4.2. Eastern equatorial Pacific Ocean

Other well studied upper oceanic crust sites that have been considered for CO<sub>2</sub> storage [Slagle and Goldberg, 2011] are in the eastern equatorial Pacific Ocean (eePO), specifically on the NE flank of the Cocos Ridge, and on the eastern flank of the East Pacific Rise (EPR) (Figure 2.13). Since 1970, numerous expeditions have provided a detailed characterisation of the extrusive section of this area (DSDP Leg 9, 16, 69, 70, 83, 92; ODP Leg 111, 137, 140, 148, 206; IODP Exp 309, 312, 335) [Alt and Teagle,

1999; *Alt et al.*, 2010; *Alt et al.*, 1996; *Becker et al.*, 1988; *Cann et al.*, 1983a; *Davis et al.*, 2004; *Teagle et al.*, 2012; *Teagle et al.*, 2006; *Tominaga et al.*, 2009; *Walther*, 2003; *Wilson et al.*, 2003; *Wilson et al.*, 2006a]. The mid-ocean ridge basalts (MORB) in the eePO, extending in age from 0 to 26 Ma, are the result of several spreading events in a region with relatively simple tectonics [*Meschede and Barckhausen*, 2000]. The thermodynamic evaluation of this region focuses on the Sites 504, 896, and 1256 (Figure 2.13 and 2.14).

Sites 504 and 896, best represented by Holes 504B and 896A respectively, are located in ~7 Ma-old crust, at a water depth of 3460 m below sea-level, and are sited ~200 km south of the intermediate spreading rate Costa Rica Rift, the eastern segment of the Cocos-Nazca spreading Ridge (summary in Table 2.4) [*Alt et al.*, 1993; 1996; *Wilson et al.*, 2003]. Hole 896A was cored only to 469 mbsf, whereas 504B is the second deepest scientific hole drilled into oceanic crust at 2111 mbsf, coring most of the way through the sheeted dike complex.

Site 1256 formed ~15 Ma ago at the East Pacific Rise (EPR) during an episode of superfast oceanic spreading (full rate >20 cm/yr) [*Wilson*, 1996]. This site is located ~1100 km east of the EPR in 3645 m of water and Hole 1256D is 1522 m deep. Together with Hole 504B, it epitomises the most complete penetration into intact upper oceanic crust (summary in Table 2.4) [*Teagle et al.*, 2012; *Teagle et al.*, 2006; *Wilson et al.*, 2003; *Wilson et al.*, 2006a].

At all three sites, the overlaying 180 - 275 m of sediments are mainly chert, biogenic silica and carbonate [*Alt et al.*, 1996; *Wilson et al.*, 2003]. Also, ODP/IODP core porosity data and downhole measurements from the three extrusive sections highlight a similar porosity, which is on average 6-8% [*Becker et al.*, 2004; *Carlson*, 2010].

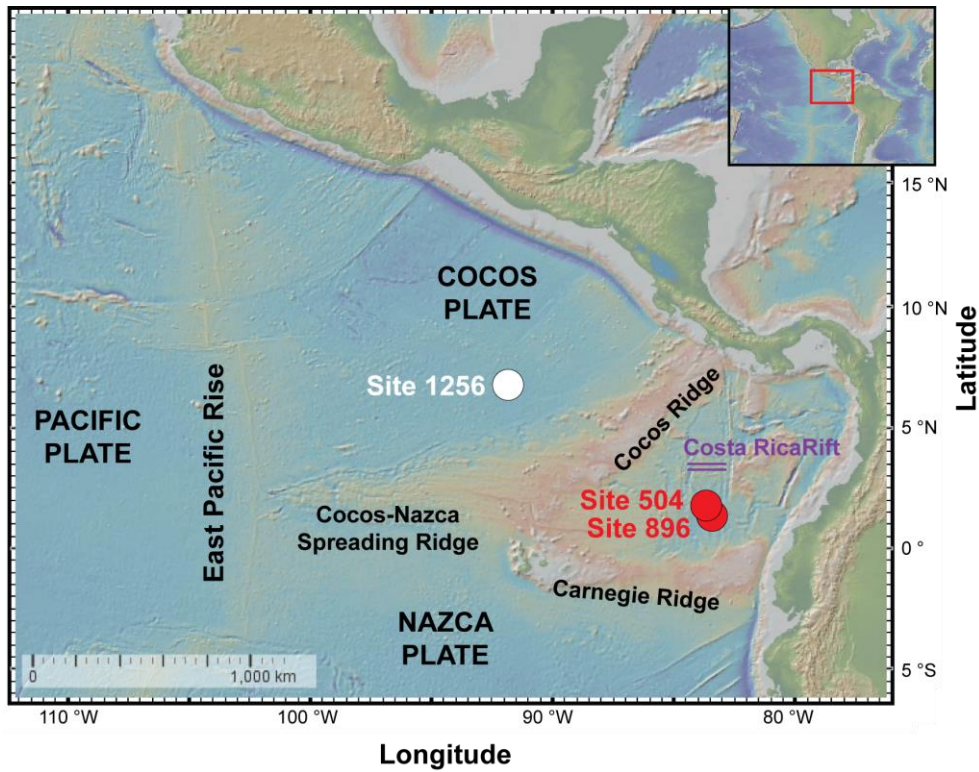


Figure 2.13: Tectonic setting of the eastern equatorial Pacific Ocean. The Sites of interest are shown in red (Site 504 and 896), and in white (Site 1256).

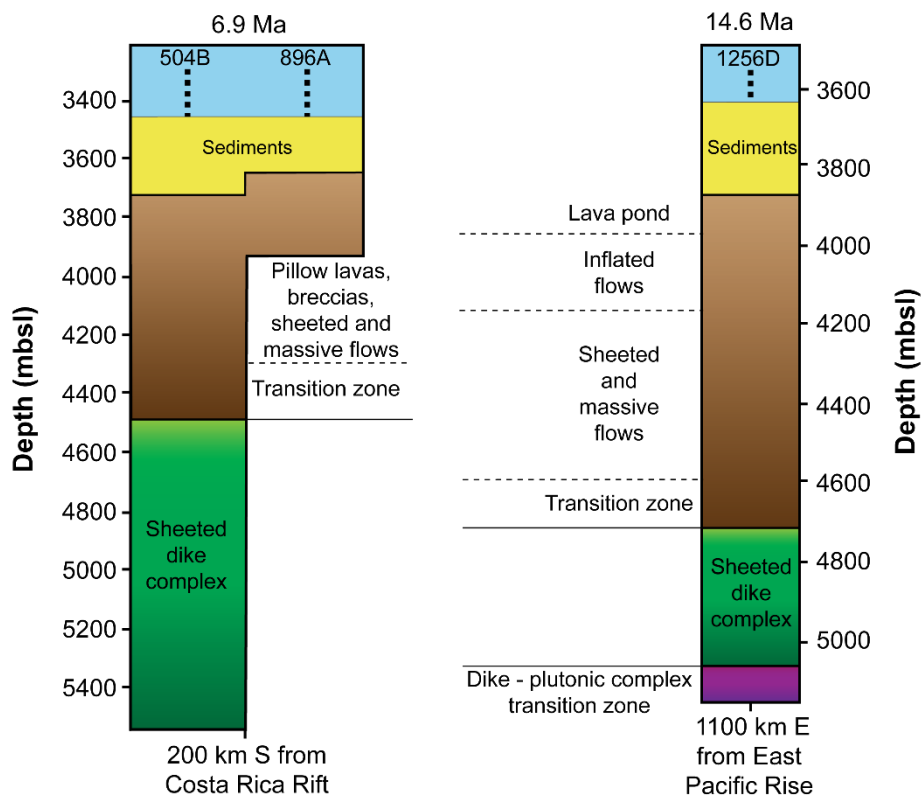


Figure 2.14: Schematic cross sections of the Holes 504B and 896A on the left (modified from Alt et al. [1996]), and Hole 1256D on the right (modified from Teagle et al. [2012]). Drill site locations and inferred basement age are shown at the bottom and top of each section, respectively.



## Temperature

The long history of temperature profiles logged during more than 30 years of expeditions to these three sites has produced important datasets [Becker *et al.*, 2004; Teagle *et al.*, 2006]. However, down-welling cold seawater flow in open-hole conditions has often influenced the temperature. Hence, only measurements at equilibrium conditions, and recorded with a wireline installation after hole sealing, have been considered in this study.

The average temperatures at the sediment-basement interface are 45, 57, and 35 °C respectively for Sites 504 [Guerin *et al.*, 1996], 896 [Becker *et al.*, 2004] and 1256 [Teagle *et al.*, 2006] (summary in Table 2.4).

## Pressure

Assuming an average water density of 1030 kg/m<sup>3</sup>, the hydrostatic pressure at the sediment-basement interface for the Sites 504, 896, and 1256 is ~38, ~37, and ~39 MPa, respectively. As explained in Section 2.3.3, this pressure calculation is the most reasonable, but for completeness here the lithostatic-assumption outcomes are as follows (sediment density = 1400-1330 kg/m<sup>3</sup> [Cann *et al.*, 1983b; Tominaga *et al.*, 2009; Wilson *et al.*, 2003]): ~39 MPa for Site 504, ~37 MPa for Site 896, and ~40 MPa for Site 1256 (Table 2.4).

*Table 2.4: Summary of the properties of the eePO Sites 504, 896 and 1256 drilled during several expeditions. Pressure - P (MPa) and temperature (°C) conditions are calculated at the sediment-basement interface. Distance from ridge axis, oceanic crustal age, water depth, and sediment thickness are also shown.*

| eePO Sites                    | 504  | 896  | 1256 |
|-------------------------------|------|------|------|
| Distance from ridge axis (km) | 228  | 229  | 1100 |
| Oceanic crustal age (Ma)      | 6.9  | 6.9  | 14.6 |
| Water depth (m)               | 3474 | 3459 | 3635 |
| Sediment thickness (m)        | 275  | 179  | 250  |
| Temperature (°C)              | 45   | 57   | 35   |
| P hydrostatic (MPa)           | 37.9 | 36.8 | 39.4 |
| P lithostatic (MPa)           | 38.9 | 37.4 | 40.0 |

### 2.4.3. Summary of two case studies

The JdFP and eePO P/T data are summarised in Figure 2.15, where the contour lines indicate the density difference ( $\Delta\rho$ ) between CO<sub>2</sub> and seawater as a function of temperature and pressure, and the phase diagram of CO<sub>2</sub> is overlaid.

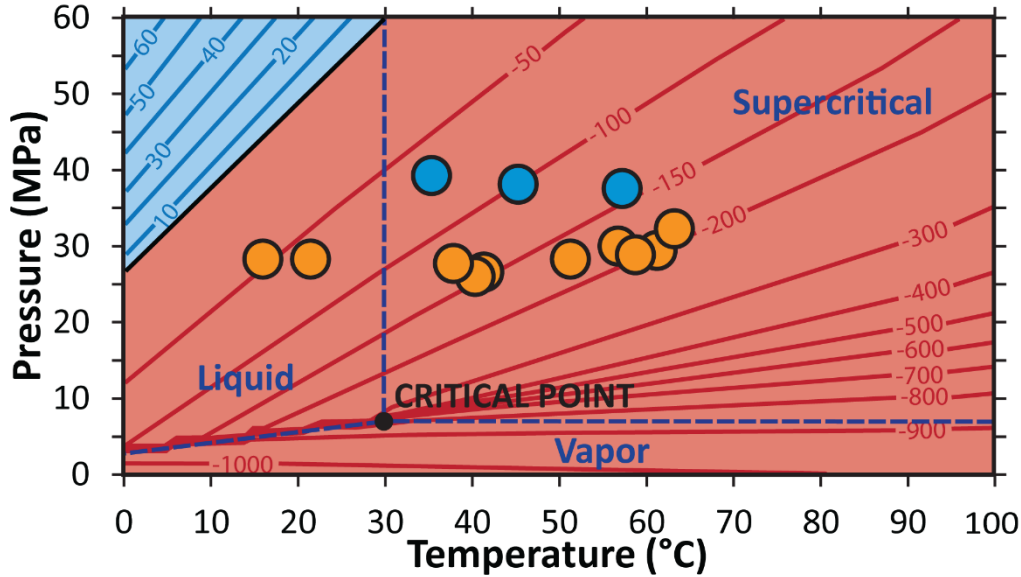


Figure 2.15: Density difference ( $\Delta\rho = \rho_{CO_2} - \rho_{seawater}$  in kg/m<sup>3</sup>) between CO<sub>2</sub> and seawater density as a function of pressure between 0 and 60 MPa, and temperature between 0 and 100 °C, with the phase diagram of CO<sub>2</sub> overlaid (modified from Marieni et al. [2013]). Positive differences ( $\rho_{CO_2}$  greater than  $\rho_{seawater}$ ) are shown in blue, negative differences in red. JdFP and eePO data are illustrated with orange and blue circles, respectively.

From this figure, it is inferred that at the sediment-basement interface:

- CO<sub>2</sub> is mostly in a supercritical state, and is liquid only where the distance from the ridge axis is <30 km;
- at all sites, CO<sub>2</sub> is less dense than seawater;
- the pressures in the eePO case are higher than in the JdFP one due to the deeper water column;
- the temperatures are similar in both regions, even though JdF Sites are younger than eePO ones.

#### 2.4.4. Validation process of temperature and pressure estimations

To validate the temperatures calculated for a world-wide scenario at the sediment-basement interface, the measured borehole temperatures at the two case-studies (JdFP and eePO) have been compared to the estimated temperature for the corresponding regions, calculated from the GDH1 model (Figure 2.16 – see Appendix CH2-6 for comparison with HSCM). The two areas have different trends of basement-sediment interface temperature as a function of age. In the eePO there is good agreement between estimated and measured temperature, whereas in the JdFP there is a discrepancy, with estimated values higher than actual measurements. The explanation lies in the sediment lithology and in the effect of the correspondent thermal conductivity ( $K_s$ ) on the temperature. The chert, biogenic silica and carbonate sediments on eePO have thermal conductivity  $\sim 1 \text{ W/(m K)}$  [Alt *et al.*, 1996; Wilson *et al.*, 2003], in agreement with the value assumed in this study with the GDH1 model. However, on the Juan de Fuca Plate, the recovered cores indicate the presence of terrigenous muddy and sandy turbiditic sediments, which have an higher measured thermal conductivity of  $\sim 2 \text{ W/(m K)}$  [Pribnow *et al.*, 2000].

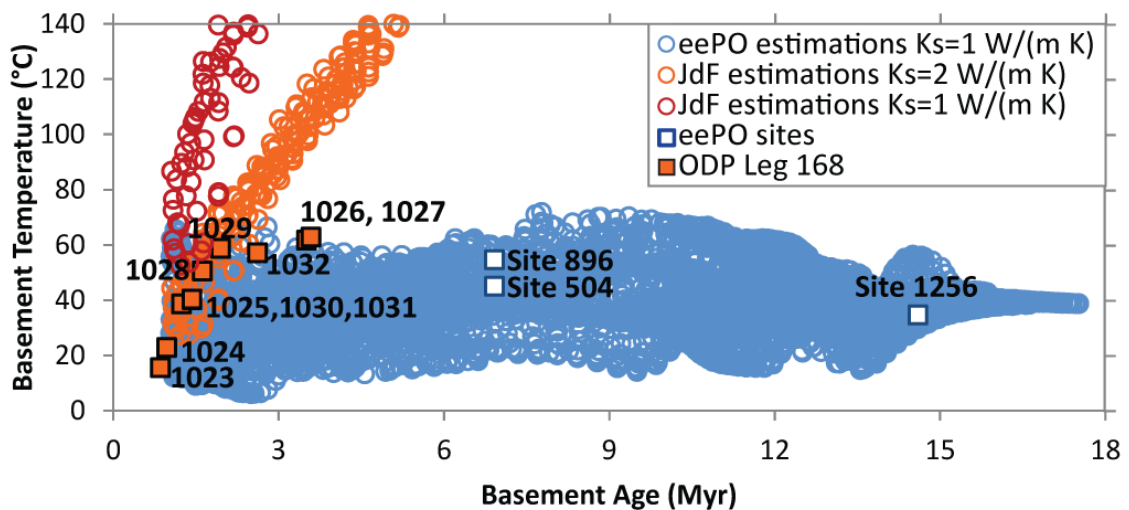


Figure 2.16: Comparison between estimated temperatures (from GDH1 model) in the eastern equatorial Pacific Ocean (eePO) and the Juan de Fuca Plate (JdFP), and measured downhole temperatures at the sediment-basement interface [Marieni *et al.*, 2013]. White squares: data from eePO [Alt *et al.*, 1993; Teagle *et al.*, 2006]; orange squares: data from JdFP [Davis *et al.*, 1997]. Circles: estimated values in the eePO (blue), and on the JdFP (red with  $K_s = 1 \text{ W/(m K)}$ ; orange with  $K_s = 2 \text{ W/(m K)}$ ).

Consequently, the temperature-age curve resulting from the recalculations for the JdFP area, using  $K_s = 2 \text{ W/(m K)}$  in the GDH1 model, shows a better fit with the majority of the temperatures at JdFP sites (Figure 2.16). On the other hand, at Sites 1026 and 1027, the measured temperatures are  $\sim 40^\circ \text{C}$  lower than predicted by the purely conductive heat flow model, and nearly identical despite the considerably different sediment thickness, which measures 229 and 614 m, respectively, (Table 2.2). This difference is justified by the local vigorous hydrothermal regime that redistributes the heat within the basement [Hutnak *et al.*, 2006; Wheat *et al.*, 2004], and is driven by very small lateral pressure gradients linked to surrounding basement outcrops (e.g., Baby Bear) [Fisher *et al.*, 2003; Spinelli and Fisher, 2004]. Also, this efficiency of lateral heat flow and hydrothermal circulation at the top of the igneous crust justifies the temperature similarity of JdF and eePO.

Hence, the temperature validation at the eePO and JdFP gives confidence in the calculations but emphasizes the need for verification of local physical properties.

Regarding the pressure estimations for the whole oceanic crust at the sediment-basement interface, no further validation is required because the hydrostatic calculations derive directly from well-constrained global datasets of sediment thickness and water depth [Divins, 2003; IOC *et al.*, 2003], and from seawater density, which show a narrow window of variations ranging from 1020 to 1050  $\text{kg/m}^3$  [Brown *et al.*, 1995]. Furthermore, the potential fluctuations in these parameters are not significant in this global context, where oceanic crust depth is on average  $\sim 3700 \text{ m}$ , which corresponds to pressures of  $\sim 37 \text{ MPa}$  [Eakins and Sharman, 2010]. In fact, a change of 100 m in sediment thickness or bathymetry produces a pressure modification of only  $\pm 0.1 \text{ MPa}$  ( $\sim 0.27\%$  of average 37 MPa), whereas the maximum change in seawater density (from 1030  $\text{kg/m}^3$ , average, to 1050  $\text{kg/m}^3$ ) yields to a maximum variation of  $+0.6 \text{ MPa}$  ( $\sim 1.6\%$  of average 37 MPa).

## 2.5. Conclusions

This chapter illustrates the phase stability of carbon dioxide in the seafloor – a first step to better understand the potential of using the oceanic crust for storing CO<sub>2</sub>. Existing global databases of crustal age, sediment thickness, and bathymetry have been investigated and combined with new studies on carbon dioxide and seawater at oceanic crust conditions. Specifically the stability of CO<sub>2</sub> has been determined at the sediment-basement interface, expressing it as a density difference between CO<sub>2</sub> and seawater.

The main findings are:

- the analysis on the thermodynamic properties of CO<sub>2</sub> and seawater highlights the importance of T, P, densities and phase diagrams in the evaluation of the overall behaviour of these two compounds;
- it is evident that carbon dioxide is denser than seawater only for pressures higher than 27 MPa (~2700 m of water) and temperature between 0 and 30 °C. It also means that within this small P-T window liquid CO<sub>2</sub> is the stable phase;
- comparison between two heat flow models confirms there are only negligible differences for the purposes of this study;
- the GDH1-based approach is a physically robust scoping study, but it presents uncertainties due to incomplete information on heat flow anomalies, local temperature irregularities, and thermal properties of sediments (i.e., thermal conductivity data);
- despite numerous holes drilled into the oceanic crust by scientific ocean drilling, the validation of the estimated temperature at the sediment-basement interface can only be assessed at thirteen sites, as they provide the only reliable drill hole temperature measurements;
- regions characterised by thermodynamically stable CO<sub>2</sub> are generally located in old oceanic crust covered by a relatively thin layer of sediments.



## Chapter 3: Oceanic crust as a reservoir for CCS

### 3.1. Introduction

Geological trapping in mafic and ultramafic rocks is a proposed mitigation strategy to sequester anthropogenic CO<sub>2</sub> from the atmosphere on a long-term basis. As described in Chapter 1, several formations have started to be considered as potential CCS reservoirs. Most of these sites are located on land. A few theoretical studies have focused on offshore basalts [Goldberg and Slagle, 2009; Goldberg *et al.*, 2008; Goldberg *et al.*, 2013; Slagle and Goldberg, 2011], which cover approximately 60% of the Earth's surface, offering huge reservoir capacity, and show evidence, such as carbonate veins, for naturally occurring CO<sub>2</sub> mineralisation [Alt and Teagle, 1999; Coggon *et al.*, 2010]. To date, no field-scale pilot projects have been developed, mainly due to uncertainties linked with scientific, technological, and economic feasibility of the sequestration method associated with oceanic reservoirs. Common questions on offshore CO<sub>2</sub> sequestration in oceanic crust can be summarised as follows:

- Can offshore basalt formations safely store CO<sub>2</sub>?
- What are the trapping mechanisms?
- Are there suitable locations for CO<sub>2</sub> sequestration in the seafloor? And if so, how accessible are these reservoirs?

So far, the global variability of sediment thickness, pressure (P), and temperature (T), and consequently the relative density ( $\rho$ ) of CO<sub>2</sub> and seawater have been considered at the sediment-basement interface of the oceanic crust (Chapter 2).

In this chapter, the oceanic crust properties described in Chapter 2, such as *in situ* density, sediment thickness, and water depth, are combined together to demonstrate the effectiveness of CO<sub>2</sub> sequestration in deep-sea basalts, and identify potential targets.

### 3.2. CO<sub>2</sub> trapping mechanisms in offshore basalts

There are three primary trapping mechanisms proposed for long-term storage of carbon dioxide in offshore basalts (Figure 3.1):

- 1- *Gravitational trapping* under pressure and temperature conditions where CO<sub>2</sub> is denser than seawater [Levine et al., 2007] – Chapter 2.
- 2- *Physical or permeability trapping*, where the presence of ≥200 m of overlying low-permeability marine sediments isolates the CO<sub>2</sub> injected into the basalts from the oceans, so that any leakage is trapped in the sediments [Goldberg and Slagle, 2009] – Chapter 3.
- 3- *Geochemical trapping*, where CO<sub>2</sub> and water react with the basalt host rocks to form geologically stable carbonate minerals, which is a process also known as “*mineral carbonation*” [Gislason et al., 2010; Seifritz, 1990] – Chapter 4.

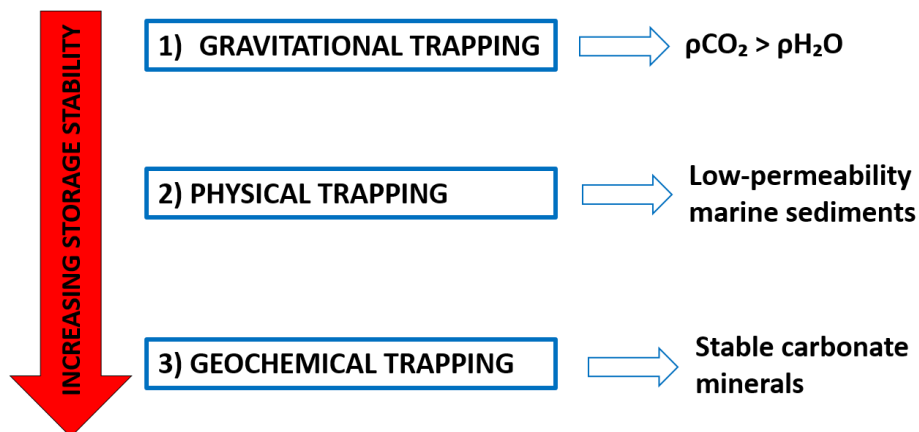


Figure 3.1: Primary trapping mechanisms for the long-term storage of carbon dioxide in offshore basalts.

Among these approaches, *carbon mineralisation* is the most permanent option (on the order of a thousand years) for CO<sub>2</sub> sequestration as the reaction products (Ca, Mg, and Fe carbonate minerals) are environmentally benign and stable, providing little possibility of accidental release and requiring little post-storage monitoring [Lackner, 2003; Lackner et al., 1995; Sipilä et al., 2008]. However, turning the carbon dioxide into a solid carbonate mineral in mafic and ultramafic rocks is not without challenges. These include



the availability of water for carbonation, the possible mobilization of trace and toxic metals, and the consequent need for low-permeability cap rocks to prevent leakage of buoyant fluids at the seafloor (*physical trapping*) [Oelkers *et al.*, 2008]. Because CO<sub>2</sub> is generally less dense than water, buoyancy tends to drive CO<sub>2</sub> upwards back to the surface [Oelkers and Cole, 2008], although, under the right conditions, CO<sub>2</sub> can be dense enough to naturally stay within the basement rocks [Levine *et al.*, 2007] (Chapter 2). Hence, in offshore conditions, the theoretical maximum storage stability is reached with the combination of all three trapping mechanisms.

An additional trapping mechanism is *solubility trapping*. This involves the dissolution of CO<sub>2</sub> into formation water to create a CO<sub>2</sub>-rich fluid that is slightly denser than the original CO<sub>2</sub>-free water [Gislason and Oelkers, 2014; Gunter *et al.*, 1993]. As for *gravitational trapping*, solubility trapping eliminates the buoyancy that drives free CO<sub>2</sub> upwards with respect to the fluid. However, the solubility approach involves i) very large quantities of seawater, and ii) the presence of porous rocks to prevent over pressurisation of the reservoir with the huge amount of pumped water. The dissolution of one tonne of CO<sub>2</sub> at a partial pressure of ~2 MPa and 25°C requires ~27 t of fresh water [Oelkers *et al.*, 2008], or 31 t of seawater [Gislason *et al.*, 2010], whereas at ~25 MPa and 20°C the required mass of seawater is ~15 t [Duan *et al.*, 2006]. The process of dissolving CO<sub>2</sub> in solution prior to its injection would also increase the storage costs [Gislason and Oelkers, 2014]. Hence, solubility trapping is not considered viable for the geological storage of CO<sub>2</sub> in deep-sea basalts.

In the next sections, the most stable conditions for combined physical and gravitational trapping of CO<sub>2</sub> in deep-sea basalts are discussed. The rock dissolution reactions involved in geochemical trapping are considered in Chapter 4.

### 3.3. Gravitational trapping

As illustrated in Chapter 2, if  $\text{CO}_2$  is denser than seawater ( $\rho_{\text{CO}_2} > \rho_{\text{seawater}}$ ), it will tend to sink into the oceanic crust, becoming gravitationally trapped. This condition is met within a narrow P-T window, with hydrostatic pressures higher than 27 MPa (>2700m of water) and temperatures between 0 and 30°C (Figure 3.2).

The map of the positive density difference ( $\Delta\rho > 0 \text{ kg/m}^3$ ) between  $\text{CO}_2$  and seawater at the sediment-basement interface, calculated using the GDH1 heat flow model, highlights how areas with suitable P-T conditions for  $\text{CO}_2$  storage are found in old oceanic crust and in the deep ocean (Figure 3.3). Most of the ridge flanks or sites close to continental margins show negative  $\Delta\rho$  conditions, due to high heat flow and thick piles of thermally isolating sediments, respectively.

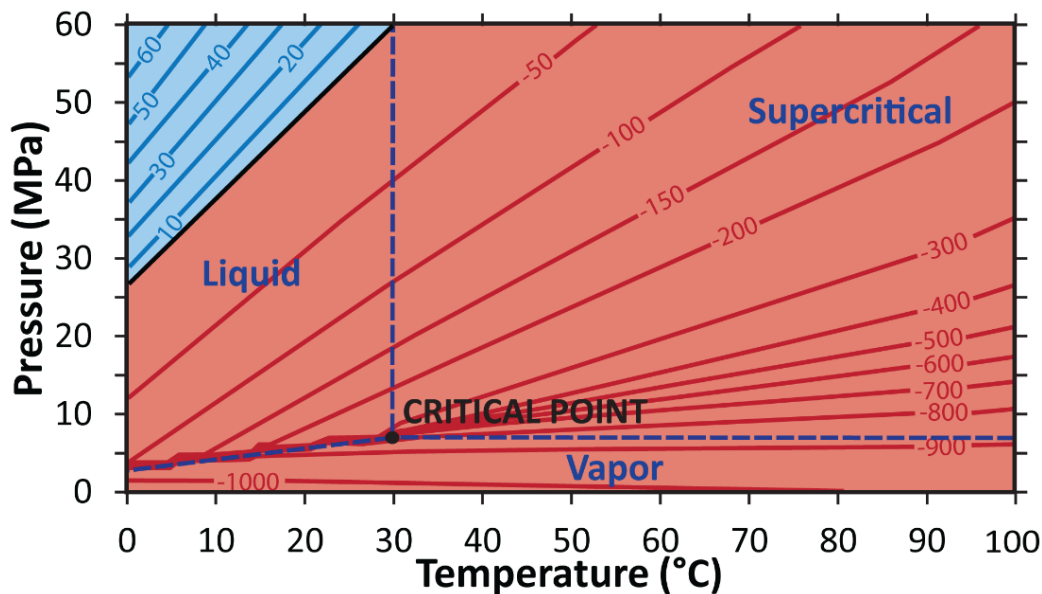


Figure 3.2: Density difference ( $\Delta\rho = \rho_{\text{CO}_2} - \rho_{\text{seawater}}$  in  $\text{kg/m}^3$ ) between  $\text{CO}_2$  and seawater density as a function of pressure between 0 and 60 MPa, and temperature between 0 and 100 °C, with the phase diagram of  $\text{CO}_2$  overlaid. Positive differences shown in blue indicate conditions for gravitational trapping [Marieni et al., 2013]. Modified version of Figure 2.15 in previous chapter.

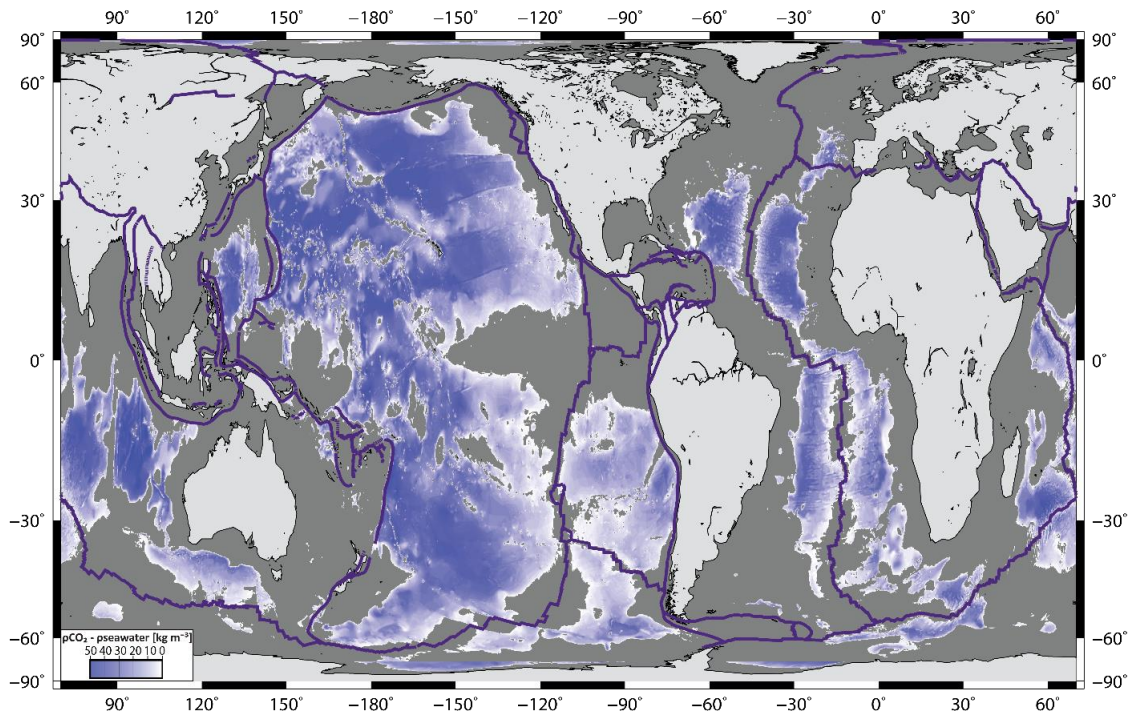


Figure 3.3: A cylindrical equal area map showing locations for gravitational trapping of CO<sub>2</sub>, using the GDH1 model for heat flow estimations. Shading shows the positive difference in density between CO<sub>2</sub> and seawater ( $\Delta\rho > 0 \text{ kg/m}^3$ ).

### 3.4. Physical trapping

The physical trapping mechanism proposed by *Goldberg and Slagle* [2009] involves the presence of  $\geq 200 \text{ m}$  of overlying sediments to isolate the injected CO<sub>2</sub> in the basalts from the oceans, and to capture any leakage in the sediment pile. This retaining strategy is based on the low permeability of the sediments, which is the parameter that controls the extent of fluid flow through the sediments. In marine sediments, permeability is low, typically ranging from  $10^{-19}$  to  $10^{-12} \text{ m}^2$ , and depends on sediment type and porosity (Table 3.1) [*Bryant and Rack*, 1990; *Spinelli et al.*, 2004]. Oceanic sediments can be divided into two main types: terrigenous (47 vol%) and pelagic (53 vol%); in turn, the pelagic sediments can be classified as calcareous (74 vol%), red clays (19 vol%), and biogenic siliceous (7 vol%) [*Hay et al.*, 1988]. In each group, permeability decreases with decreasing porosity, which decreases with depth. The permeability also has the tendency to increase with grain size; for example higher values are recorded in the sandy layers of turbidites (coarsest) than in red clays (finest).

Table 3.1: Oceanic sediment types (data from Hay et al. [1988]) and relative porosity and permeability (data from Spinelli et al. [2004]).

| Oceanic sediments |             |                       | Porosity | Permeability                          |
|-------------------|-------------|-----------------------|----------|---------------------------------------|
|                   |             |                       | %        | m <sup>2</sup>                        |
| 47%               | Terrigenous |                       | 20 - 85  | 10 <sup>-19</sup> - 10 <sup>-13</sup> |
| 53%               | Pelagic     | Calcareous 74%        | 50 - 85  | 10 <sup>-17</sup> - 10 <sup>-13</sup> |
|                   |             | Red clays 19%         | 60 - 90  | 10 <sup>-18</sup> - 10 <sup>-13</sup> |
|                   |             | Biogenic siliceous 7% | 45 - 90  | 10 <sup>-19</sup> - 10 <sup>-12</sup> |

The NOAA-gridfive sediment database [Divins, 2003] has been used to determine the sediment thickness for all the areas with underlying oceanic crust (Appendix CH2-4). In general, the global distribution of sediment thickness is a function of oceanic lithosphere age [Müller et al., 1997; Spinelli et al., 2004]. If pairs of sediment thickness and lithospheric age values are grouped by age into 5 Ma bins, most of the bin-median values lie below 700 m and above 200 m (Figure 3.4). The average sediment thickness calculated from Divins [2003] database is ~400 m.

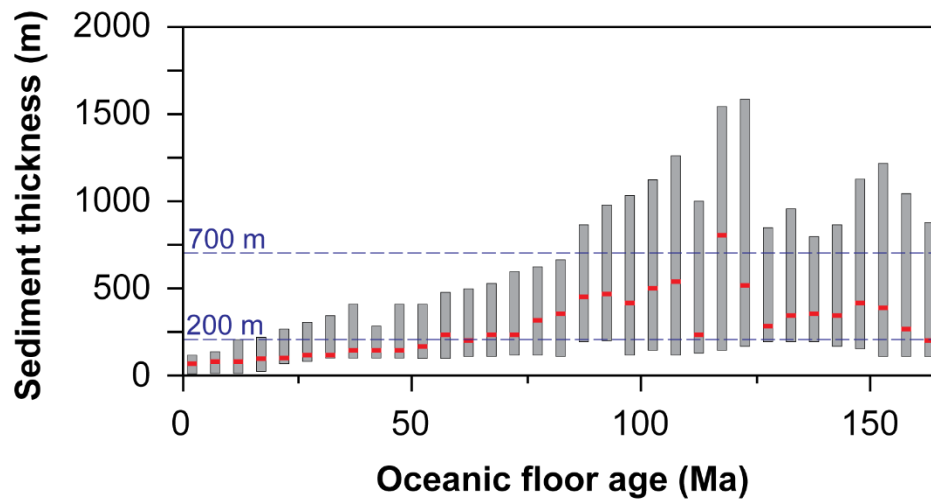


Figure 3.4: Sediment thickness (data from Divins [2003]) versus oceanic floor age (data from [Müller et al., 1997]) – modified from Spinelli et al. [2004]. Pairs of sediment thickness and lithospheric age values are determined on a grid with a spacing of 1.67° x 1.67°, then grouped by age into 5 Myr bins. The box plot for each 5-Myr age bin shows the median sediment thickness (red horizontal line), the 25th and 75th percentile (bottom and top of the box, respectively), the 200 and 700 m limits (blue dashed lines).

In this study, based on the global distribution of sediment thickness, a minimum of 200 m has been chosen as the lowest physical trapping constraint to ensure a continuous low permeability blanket over minor basement topography such as faulted ridges or seamounts that might puncture the sediment cover and allow the ingress or egress of basement fluids [Divins, 2003]. This is consistent with recommendations for CO<sub>2</sub> storage in deep saline aquifers that require a caprock thickness of >100 m [e.g. Chadwick et al., 2008].

To estimate the maximum sediment thickness, the density difference has been calculated for a wide range of lithospheric ages and sediment thicknesses using the GDH1 model for both water depth and heat flow, and assuming a hydrostatic sediment column (Figure 3.5). Based on global average conditions, GDH1 indicates a restricted zone where gravitational trapping is possible, and that anywhere with more than ~600 m of sediments CO<sub>2</sub> is likely to be gravitationally unstable due to the high temperatures. Using the HSCM (see Appendix CH3-1), the equivalent limit is ~1000 m. Hence, the upper sediment thickness limit has been set at 700 m.

From these constraints, a global map for the physical trapping strategy can be plotted for areas where sediment thicknesses is between 200 and 700 m (Figure 3.6).

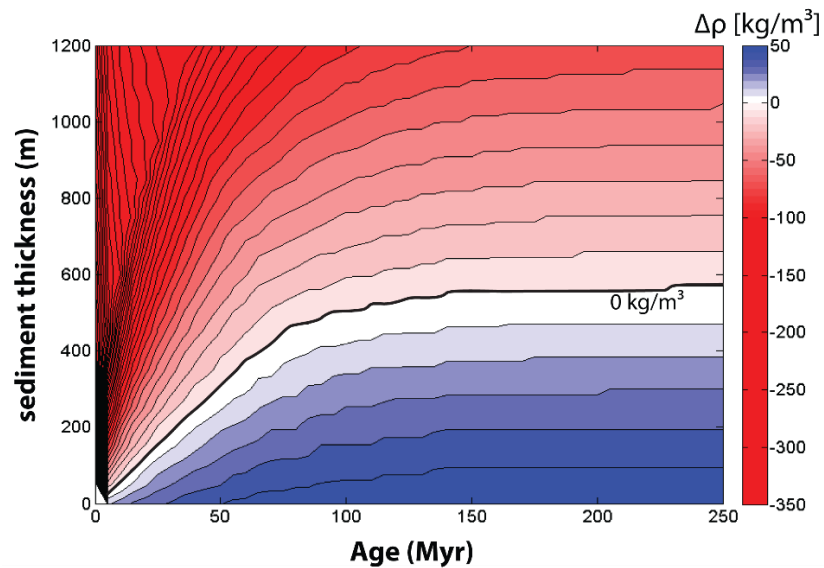


Figure 3.5: Density difference between CO<sub>2</sub> and seawater at the sediment-basement interface as a function of plate age and sediment thickness using the GDH1 model to determine both water depth and thermal conditions [Marieni et al., 2013]. Sediment thicknesses below the heavy black line show where positive density differences required for stable gravitational trapping are achieved.



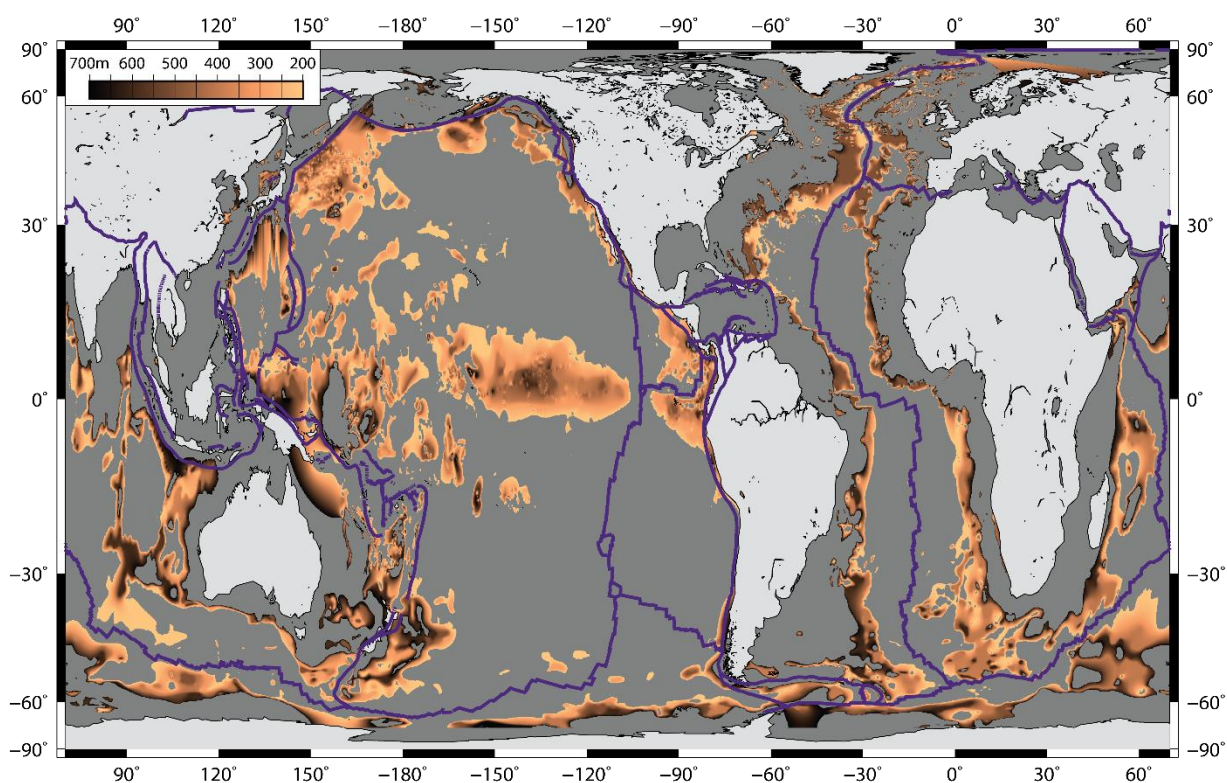


Figure 3.6: A cylindrical equal area map showing locations for physical trapping of CO<sub>2</sub> in the seafloor. Shading shows the sediment thickness between 200 and 700 m.

### 3.5. Offshore basalt targets

The optimal conditions for CO<sub>2</sub> storage in deep-sea basalts are reached by combining gravitational and physical trapping. The results from the individual analysis of each trapping mechanism have been merged in a global map (6 minute resolution), and expressed as density difference at the sediment-basement interface (Figure 3.7). A positive density difference identifies locations where 1) CO<sub>2</sub> is denser than seawater at the sediment-basement interface ( $\rho_{\text{CO}_2} > \rho_{\text{seawater}}$ ) (Figure 3.3), and 2) the sediment thickness is between 200 m and 700 m (Figure 3.6). The global map produced using the HSCM is available in the Appendix CH3-2.

Much of the upper oceanic crust is not suitable for the geological sequestration of CO<sub>2</sub> by gravitational and physical trapping. However, five potential targets can be identified based on the positive  $\Delta\rho$  between CO<sub>2</sub> and seawater, the oceanic crustal age, the

sediment thickness, and the distance to major industrial CO<sub>2</sub> sources [*International Energy Agency Greenhouse Gas R&D Programme*, 2002]. All the targets are in water depths >5000 m. Selected reservoirs are in the Indian Ocean between Indonesia and Australia (inset **a**); in the northwest Pacific Ocean near the east coast of Japan and Russia (inset **b**), and south of the Aleutian Islands (inset **c**); and in the Atlantic Ocean near Bermuda (inset **d**) and close to South Africa (inset **e**) (Table 3.2). Other areas also have suitable conditions for carbon dioxide trapping, but these options have not yet been explored due to their smaller sizes and lower  $\Delta p$ , although some are closer to land. Potential storage volume for each target has been computed (Table 3.2), assuming an average porosity of 10% [*Carlson and Herrick*, 1990; *Johnson and Pruis*, 2003], even for old oceanic crust (e.g., ODP Hole 801C [*Jarrard et al.*, 2003]), and 300 m as a reasonable thickness of permeable pillow lavas for old crust. The storage capacity in each area is between ~13,800 and 127,800 Gt of CO<sub>2</sub>. Given the ~1840 Gt of cumulative anthropogenic CO<sub>2</sub> emissions in the atmosphere since the Industrial Revolution [*Allen et al.*, 2009], and at the current global annual anthropogenic flux of ~36 Gt of CO<sub>2</sub> per year [*Le Quéré et al.*, 2015], even the smallest identified reservoir (inset **c**) could provide sufficient carbon dioxide sequestration capacity to accommodate past CO<sub>2</sub> emissions and also several centuries worth of future emissions (Figure 3.7).

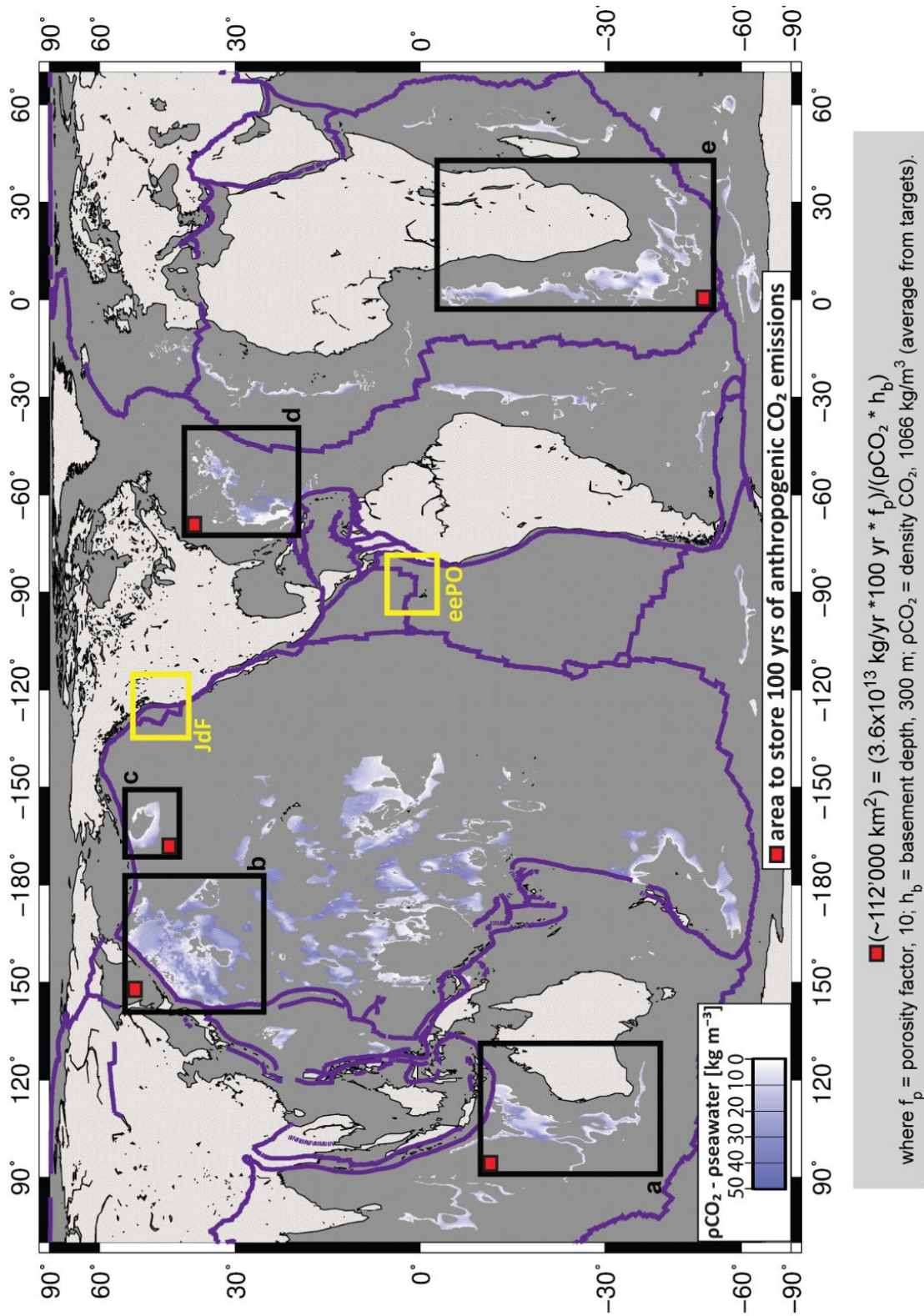


Figure 3.7: A cylindrical equal area map showing locations for stable geological sequestration of  $\text{CO}_2$  [modified from Marieni et al., 2013]. Shading shows the difference in density between  $\text{CO}_2$  and seawater in areas where the sediment thickness is between 200 and 700 m and the  $\text{CO}_2$  is denser than seawater. Five potential reservoirs (insets a–e) have been identified. As described in the equation at the bottom, the red box indicates the area required to store 100 yrs of current anthropogenic emissions of  $\text{CO}_2$  (36 Gt/yr), assuming a pillow lava thickness of 300 m and 10% porosity [Carlson and Herrick, 1990; Jarrard et al., 2003; Johnson and Pruis, 2003]. Yellow boxes show regions in Figure 3.11, previously discussed in Chapter 2.



Table 3.2: Properties of the five potential reservoirs.  $\Delta\rho$ , in situ excess density of  $\text{CO}_2$  over seawater; age from [International Energy Agency Greenhouse Gas R&D Programme, 2002; Müller et al., 2008]; sediment thickness from the NOAA database [Divins, 2003]. Distance of the reservoir from land is taken from the nearest stationary source of  $\text{CO}_2$  according to the IEA GHG database [International Energy Agency Greenhouse Gas R&D Programme, 2002].

| Code | Location     | Area<br>[ $\times 10^6 \text{ km}^2$ ] | Pore<br>volume<br>[ $\times 10^4 \text{ km}^3$ ] | $\rho\text{CO}_2$<br>[ $\text{kg/m}^3$ ] | $\text{CO}_2$<br>[Gt] | $\Delta\rho$<br>[ $\text{kg/m}^3$ ] | Age<br>[Ma] | Sediment<br>thickness<br>[m] | Distance<br>[km] |
|------|--------------|----------------------------------------|--------------------------------------------------|------------------------------------------|-----------------------|-------------------------------------|-------------|------------------------------|------------------|
| a    | Indian Ocean | 1.47                                   | 4.42                                             | 1066                                     | 47,162                | 18                                  | 85          | 335                          | 1500             |
| b    | NW Pacific   | 3.97                                   | 11.9                                             | 1073                                     | 127,870               | 24                                  | 100         | 310                          | 1300             |
| c    | S-Aleutians  | 0.43                                   | 1.30                                             | 1063                                     | 13,791                | 15                                  | 60          | 275                          | 1500             |
| d    | Bermuda      | 1.15                                   | 3.45                                             | 1066                                     | 36,780                | 17                                  | 80          | 320                          | 1500             |
| e    | SE Atlantic  | 2.22                                   | 6.66                                             | 1062                                     | 70,701                | 14                                  | 85          | 290                          | 1700             |

To confirm the assumption of 300 m as reasonable thickness of permeable old basalts, pressures and temperatures (using GDH1 model, with 2 W/(m K) as thermal conductivity of pillow lavas [Clauser, 2006]) have been calculated at 300 m into the basement for all the gravitationally and physically suitable locations. The density difference results are illustrated on a 6°×6° global map (Figure 3.8).

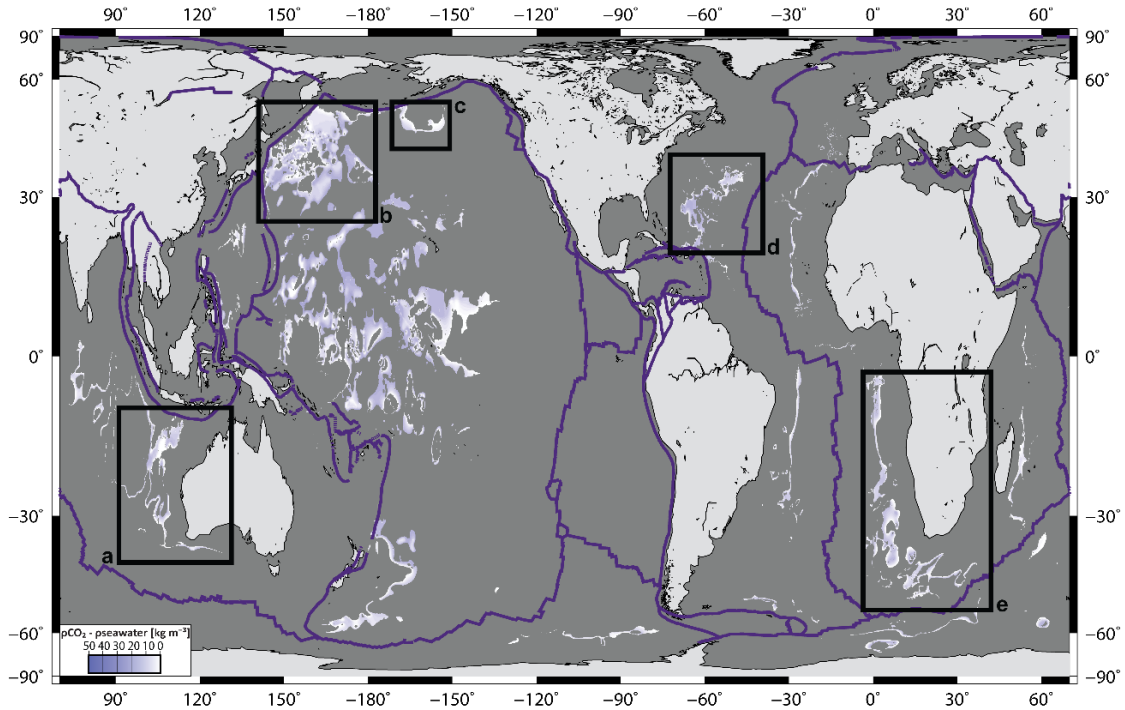


Figure 3.8: A cylindrical equal area map showing locations for stable geological sequestration of  $\text{CO}_2$  at 300 m into the basement. Shading shows the difference in density between  $\text{CO}_2$  and seawater in areas where the sediment thickness is between 200 and 700 m and the  $\text{CO}_2$  is denser than seawater. 71% (by surface area) of the areas identified in Figure 3.7 are valid at this depth into the basement.

The size of the areas identified in these calculations are summarised in Table 3.3, expressed as a function of the total oceanic floor area, which is  $\sim 2.96 \times 10^8 \text{ km}^2$  [Müller et al., 2008]. If only physical trapping with sediment thickness between 200 and 700 m is considered at the sediment-basement interface (**1** in Table 3.3), the area of suitable oceanic crust for CO<sub>2</sub> storage is 31.5% of the total, and includes the sites where  $\Delta\rho < 0 \text{ kg/m}^3$  (seawater is denser than CO<sub>2</sub>). If only gravitational trapping with  $\Delta\rho > 0 \text{ kg/m}^3$  (CO<sub>2</sub> is denser than seawater) is taken into account, the recommended locations cover 48.1% of the seafloor, with an average  $\Delta\rho$  of  $\sim 26 \text{ kg/m}^3$  (**2** in Table 3.3). However, when both gravitational and physical trapping of CO<sub>2</sub> are combined, this area falls to only 7.5% at the sediment-basement interface (**3a** in Table 3.3) and 5.3% at 300 m into the basaltic crust (**3b** in Table 3.3), which is 71% of the area at the sediment-basement interface. Careful consideration of the behaviour of CO<sub>2</sub> in the crust is essential in selecting a suitable area for CO<sub>2</sub> storage.

Table 3.3: Surface area estimations as a function of type of trapping and sediment thickness. The relative area is calculated as a percentage of all the oceanic floor area ( $2.96 \times 10^8 \text{ km}^2$ ) estimated with age [Müller et al., 2008].

| Code | Type of trapping         | Sediment thickness<br>[m] | Location                    | Average $\Delta\rho$<br>[kg/m <sup>3</sup> ] | Area<br>[km <sup>2</sup> ] | Relative area |
|------|--------------------------|---------------------------|-----------------------------|----------------------------------------------|----------------------------|---------------|
| 1    | physical                 | 200 - 700                 | sediment-basement interface |                                              | $9.32 \times 10^7$         | 31.5%         |
| 2    | gravitational            |                           | sediment-basement interface | 26                                           | $1.42 \times 10^8$         | 48.1%         |
| 3a   | gravitational + physical | 200 - 700                 | sediment-basement interface | 19                                           | $2.23 \times 10^7$         | 7.5%          |
| 3b   | gravitational + physical | 200 - 700                 | 300 m into basement         | 11                                           | $1.57 \times 10^7$         | 5.3%          |
| 4    | gravitational + physical | 200 - 500                 | sediment-basement interface | 19                                           | $2.19 \times 10^7$         | 7.4%          |
| 5    | gravitational + physical | 0 - 200                   | sediment-basement interface | 28                                           | $1.17 \times 10^8$         | 39.4%         |
| 6    | gravitational + physical | 100-200                   | sediment-basement interface | 29                                           | $5.07 \times 10^7$         | 17.1%         |
| 7    | gravitational + physical | 0 - 500                   | sediment-basement interface | 27                                           | $1.42 \times 10^8$         | 48.0%         |
| 8    | gravitational + physical | 0 - 700                   | sediment-basement interface | 26                                           | $1.42 \times 10^8$         | 48.1%         |

These calculations have all assumed a sediment thickness between 200 and 700 m (Section 3.4). However, in areas where the CO<sub>2</sub> is gravitationally stable and basement topography is low, 200 m of sediments may not be required to physically trap the CO<sub>2</sub> as theoretically the CO<sub>2</sub> would remain in the basalt even with no sediment cover. In order to better explore the effect of sediment thickness, five different scenarios have been considered (Table 3.3 and relative global maps in Appendices CH3-3 A, B, C, D, E): **4-** from 200 to 500 m; **5-** from 0 to 200 m; **6-** from 100 to 200 m; **7-** from 0 to 500 m; **8-** from 0 to 700 m. The results highlight the way in which the extent of CO<sub>2</sub> offshore reservoirs could be improved from 7.5 to 39.4% of the ocean floor if  $\leq 200$  m of marine sediments blanketed the gravitationally stable areas ( $\Delta\rho$  of  $\sim 28$  kg/m<sup>3</sup>). Also, almost half of these locations are characterised by sediment thickness ranging from 100 to 200 m ( $\Delta\rho$  of  $\sim 29$  kg/m<sup>3</sup>). Increasing the upper limit of sediment thickness from 500 to 700 m in gravitationally stable regions adds only 0.1% to the relative available area, and so increasing sediment thickness is relatively unimportant.

These evaluations, based on global data sets, show that CCS using subsea basalts as the storage medium has considerable potential, with many alternative storage conditions. However, regional investigations are required to determine local sediment properties, thicknesses, continuity, and seafloor thermal gradients. Drilling to facilitate detailed lithological, physical, thermal, and hydrological characterization of the sediment overburden and target basalt formations is essential.

In the next Section, case studies of unsuitable reservoirs (JdF and eePO in Figure 3.7), and an example of local investigations in the offshore region to the west of the Exmouth Plateau in NW Australia (*target a* in Figure 3.7) have been undertaken.

### 3.5.1. JdF and eePO as unsuitable CCS targets

As described in Section 2.4, the Juan de Fuca Plate (JdFP) and eastern equatorial Pacific Ocean (eePO) are two regions where the physical, geochemical, and structural properties of the upper oceanic crust are well characterised. Both regions have previously been proposed as potential offshore targets for carbon dioxide sequestration [Goldberg *et al.*, 2008; Slagle and Goldberg, 2011]. They have thick sediment layers on top of the basement, and show relatively high porosity (~10%) and permeability values ( $10^{-17} - 10^{-13} \text{ m}^2$ ) in the upper 300 m of the crust [Anderson *et al.*, 1985; Becker, 1996; Becker and Fisher, 2000; Davis and Becker, 1998]. The JdFP has the advantage of being located at short pipeline distances to populated areas and CO<sub>2</sub> sources. In the evaluation of these sites, the advantages of long-term carbon sequestration via mineral carbonation, physical and gravitational trapping in oceanic crust were discussed. *Gravitational trapping* has been described for water depths greater than 2700 m, where injected CO<sub>2</sub> was assumed to be denser than seawater on the basis of House *et al.* [2006] density calculations. This trapping approach was supported by the outcomes of direct experiments on the ocean disposal of CO<sub>2</sub> carried out by Brewer *et al.* [1999], where CO<sub>2</sub> hydrate (~6 H<sub>2</sub>O+CO<sub>2</sub>) was forming from the injection of liquid CO<sub>2</sub> in the water depth range 2700 to 4500 m. The crystalline CO<sub>2</sub> hydrate is denser but less soluble than liquid CO<sub>2</sub> in seawater <2 °C, and its formation impedes the upward flow of the underlying CO<sub>2</sub> (liquid). However, in these previous studies on JdFP and eePO the phase relationships at the sediment-basement interface were not taken into account. The thick sediment blanket covering young oceanic crust results in high temperatures at the sediment-basement interface. Under these conditions CO<sub>2</sub> is a supercritical fluid, and consequently much less dense than seawater. Hence, contrary to previous suggestions, sites on the Juan de Fuca Plate and in the eastern equatorial Pacific Ocean are unsuitable for gravitational trapping of carbon dioxide (Figure 3.11) because the CO<sub>2</sub> will naturally tend to escape as a buoyant supercritical fluid.

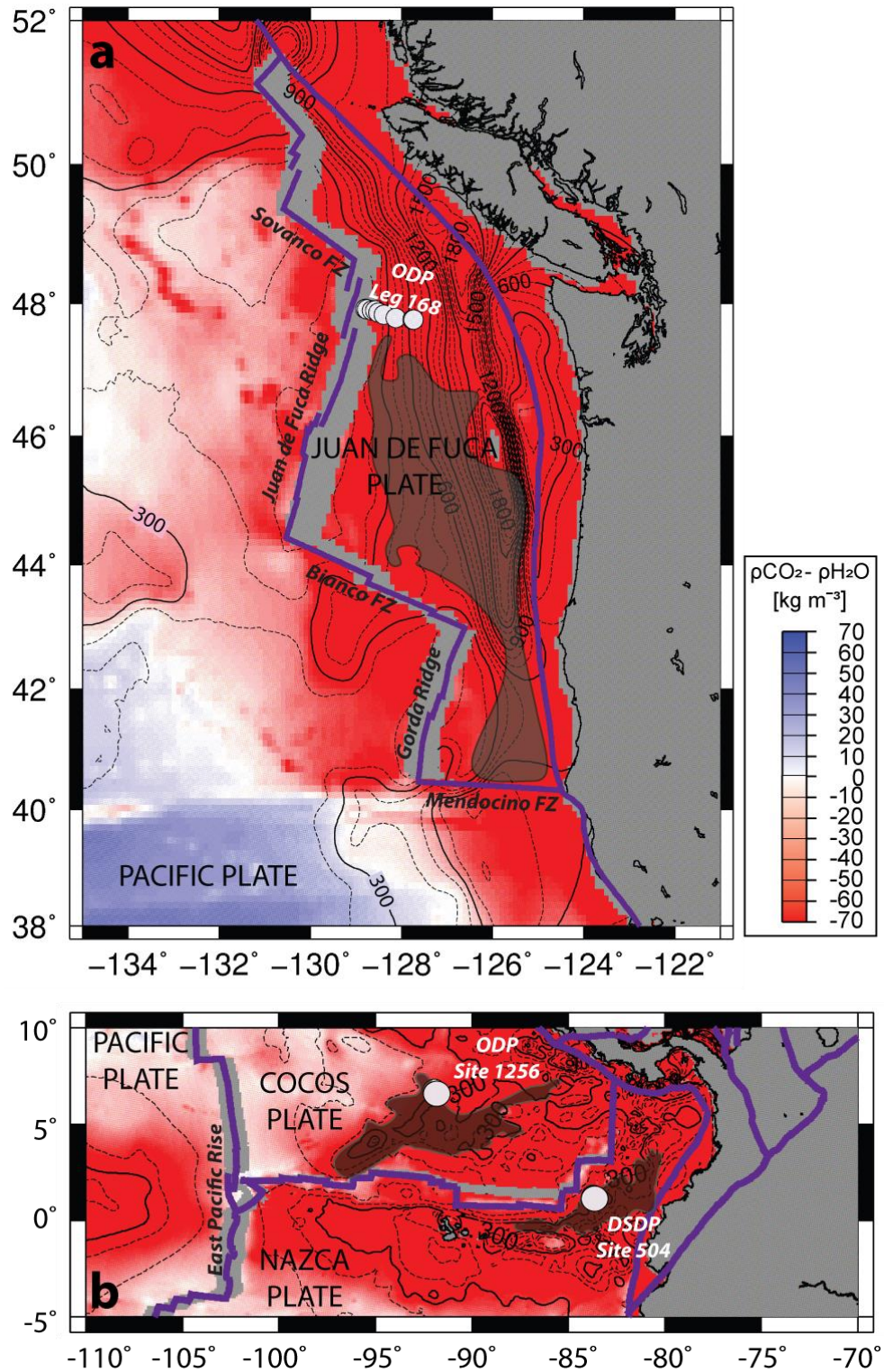


Figure 3.9: Map of density difference  $\Delta\rho$  between  $\text{CO}_2$  and seawater at (a) the Juan de Fuca Plate (with  $K_s = 2 \text{ W/(m K)}$ ) and (b) the eastern equatorial Pacific Ocean (with  $K_s = 1 \text{ W/(m K)}$ ). Sediment thicknesses [m] are shown by black contour lines. The dark shadows show the previously suggested regions for deep-sea basalt  $\text{CO}_2$  sequestration [Goldberg et al., 2008; Slagle and Goldberg, 2011].



### 3.5.2. Target a: Indian Ocean

Target **a** is located in the Indian Ocean, offshore Western Australia, between the Java Trench, the Ninetyeast Ridge, Broken Ridge, and the Diamantina Fracture Zone (Figure 3.10). This zone of the Indian Ocean is part of the Australian Plate, and is one of the oldest continent / ocean boundaries on Earth (~160 Ma) [Gradstein *et al.*, 1990]. The Exmouth plateau (Figure 3.10) is a major offshore petroleum province (e.g. Shell, and Woodside). This subsided continental platform is located ~300 km from the NW coast of Australia, and comprises ~10 km block-faulted Palaeozoic to Mesozoic (542-65 Ma) sedimentary rocks [Exon and Willcox, 1978]. The abundance of hydrocarbons in the sandstone formations of the area makes *target a* an economically interesting location to start more detailed investigations into the potential for deep-sea basalt CO<sub>2</sub> sequestration.

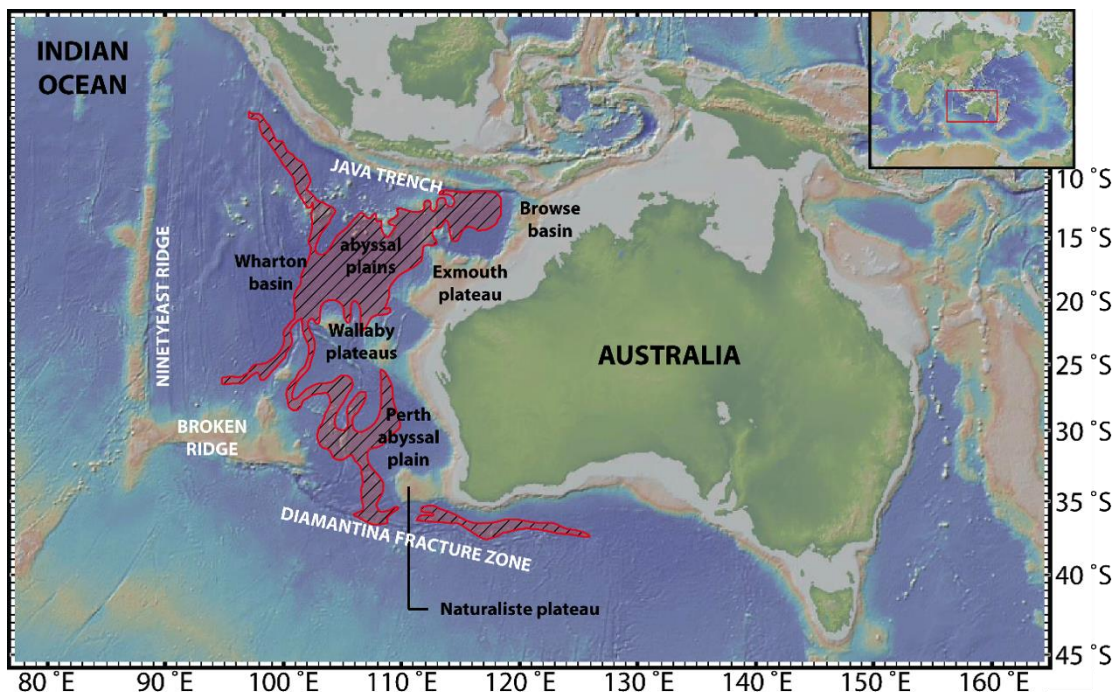


Figure 3.10: Tectonic setting of the offshore Western Australia. The Indian Ocean target “a” is highlighted in red.

Since the Jurassic (~200 Ma), many rifting events have occurred in this area, with at least three polar-wander path shifts influencing the oceanic spreading direction

[Schettino and Scotese, 2005]. As a consequence, numerous plateaus, ridges, basins, and abyssal plains characterise this region of the Indian Ocean. The area near *target a* has been the focus of five scientific ocean drilling cruises (DSDP Leg 26, 27, and 28; ODP Leg 122 and 123), which were all primarily focussed on the biostratigraphy of the sediments on top of the basement, and on the geodynamic reconstructions of the local paleogeography [Gradstein et al., 1992; Hayes et al., 1975; Luyendyk and Davies, 1974; Veevers et al., 1974; von Rad et al., 1992] (Figure 3.11). Hence, there are few data from the sediment-basement interface to characterise *target a*, and to compare the empirical measurements with the oceanic crustal age-derived estimations, which have been carried out in this study.

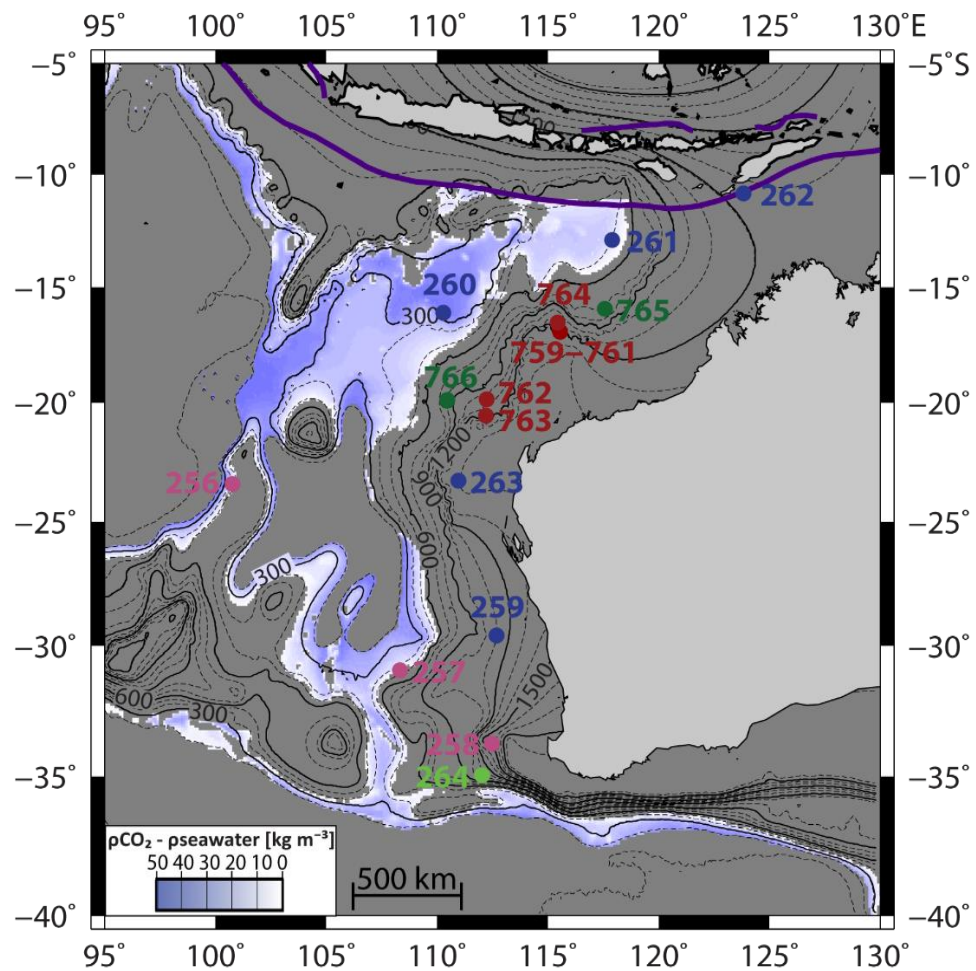


Figure 3.11: An equal area map showing the potential target *a* for CO<sub>2</sub> storage. Shading shows the difference in density between CO<sub>2</sub> and seawater in areas where the sediment thickness is between 200 and 700 m and the CO<sub>2</sub> is denser than seawater. Sediment thicknesses [m] are shown with black contour lines [Divins, 2003]. The five scientific oceanic drilling cruises are also shown on the map (DSDP Leg 26 in pink; DSDP Leg 27 in blue; DSDP Leg 28 in light green; ODP Leg 122 in red; and ODP Leg 123 in dark green).

### 3.5.2.1. Comparison between estimated and measured values

For each of the five scientific cruises, the results from drilling at representative sites are summarised in Table 3.4, together with the estimations of oceanic crustal age, water depth, sediment thickness, and temperature at the sediment-basement interface used in this study from global databases. The data relative to the ODP Leg 122 are also shown for comparison, even if the Sites are located on the continental margin.

Igneous basement was reached in three of the cruises (DSDP Leg 26, Leg 27, and ODP Leg 123), but only at one site (ODP Leg 123 – Site 765) temperature and porosity measurements were recorded [Brereton, 1992]. This was the first time a high resolution temperature tool (Temperature Logging Tool – TLT) was successfully used during scientific oceanic drilling. However, the temperature measurements at the sediment-basement interface were recorded several days after drilling through the basalts, and they were found to be lower than predicted by heat flow and thermal gradient measurements. Castillo [1992] attributed this cooling to the flow of cold seawater or formation fluids into the borehole during the open-hole conditions. The inferred temperature at the sediment-basement interface was then corrected from 26 to 37 °C using the Bullard method<sup>1</sup> [Bullard, 1947]. The crustal age of Site 765 was estimated to be 140 Ma, based on the oldest recovered sediment, but more recent age analysis using K/Ar on celadonite indicates it is 15 Myr older (155 Ma) [Gradstein and Ludden, 1992]. With this age of 155 Ma, the basalt at Site 765 is the oldest basement in the Indian Ocean.

In general, the data collected during these scientific drilling legs show the following: oceanic crustal ages on average older than 100 Ma; average thermal conductivity of ~1.3 W/(m K); bathymetry >5000 m; and sediment thicknesses between ~250 and ~1000 m (Table 3.4).

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<sup>1</sup> The Bullard method is a technique used to estimate one dimensional, steady-state conductive heat flow in a layered medium, based on the temperature gradient and the summed thermal resistance of a number of depth intervals.



Table 3.4: Summary of drilling results (grey shading) and estimations from the global databases used in this study (yellow shading) for the representative sites of each scientific cruise. Blank spaces indicate not available measurements from cruises. Estimated values that show a significant difference ( $\pm 20\%$ ) from the measurements are underlined.

| CRUISES                                         |         | DSDP<br>Leg 26 |            | DSDP<br>Leg 27 |      | DSDP<br>Leg 28 | ODP<br>Leg 122 |         | ODP<br>Leg 123 |
|-------------------------------------------------|---------|----------------|------------|----------------|------|----------------|----------------|---------|----------------|
| SITES                                           |         | 256            | 257        | 260            | 261  | 264            | 759            | 763     | 765            |
| Oceanic crustal age                             | Ma      | 103            | 109        | 129            | 153  |                |                |         | 155            |
|                                                 |         | 103            | 109        | 129            | 153  | 103            |                |         | 155            |
| Bathymetry                                      | m       | 5361           | 5278       | 5702           | 5687 | 2876           | 2092           | 1368    | 5730           |
|                                                 |         | 5268           | 5013       | 5677           | 5642 | 2821           | <u>2590</u>    | 1350    | 5713           |
| Deepest penetration                             | m       | 270            | 327        | 332            | 580  | 215            | 308            | 1037    | 1195           |
| Sediment thickness                              | m       | 251            | 262        | 323            | 533  | 171            | > 308          | > 1037  | 936            |
|                                                 |         | <u>487</u>     | <u>462</u> | 266            | 496  | <u>1040</u>    | 1237           | 1201    | <u>710</u>     |
| Drill depth into basement                       | m       | 19             | 65         | 9              | 47   | 44             |                |         | 259            |
| Sediment thermal conductivity (K <sub>s</sub> ) | W/(m K) | 0.8            | 0.9        |                |      |                | 1.0-2.2        | 1.3-1.8 | 0.9-1.7        |
|                                                 |         | 1              |            |                |      |                |                |         |                |
| Basement porosity                               | %       | 5              |            |                |      |                |                |         |                |
|                                                 |         | 10             |            |                |      |                |                |         |                |
| T at sediment-basement interface                | °C      | 26 → 37        |            |                |      |                |                |         |                |
|                                                 |         | 28             | 26         | 16             | 27   | 58             | 63             | 61      | 37             |

The estimated values of oceanic crustal age (Figure 2.1, from Müller *et al.* [2008]) and bathymetry (Appendix CH2-4, from IOC *et al.* [2003]) are in good agreement with measured values. In contrast, sediment thickness (Appendix CH2-2, from Divins [2003]) shows differences of ~200 m, and up to ~900 m in the case of Site 264 (Table 3.4). The NOAA sediment thickness database used in this study [Divins, 2003] takes into account i) previously published isopach maps (e.g. Divins and Rabinowitz [1990]; Hayes and LaBrecque [1991]; Ludwig and Houtz [1979]; Matthias *et al.* [1988]), ii) ocean drilling results, and iii) archived seismic reflection profiles. However, the data interpolation is not a perfect process, especially along continental margins. To explore the validity of sediment thickness estimations in this area, the site specific measurements of scientific

ocean drilling sites are compared to the scanned Single-Channel Seismic (SCS) reflection profiles collected on cruises of Lamont-Doherty's research vessels Robert D. Conrad (cruise RC1403), and Vema (cruise V2819) [<http://www.geomapapp.org>] (Figure 3.12). The seismic profiles show major changes in the acoustic basement depth in the space of a few kilometres for sites on or along the continental margin (e.g., Sites 759 and 765, respectively), suggesting that the sediment thickness discrepancies between measurements and estimations are the result of local variability. However, relatively constant sediment thickness observed around Site 260 confirms the validity of global estimations for mid-ocean ridge flanks.

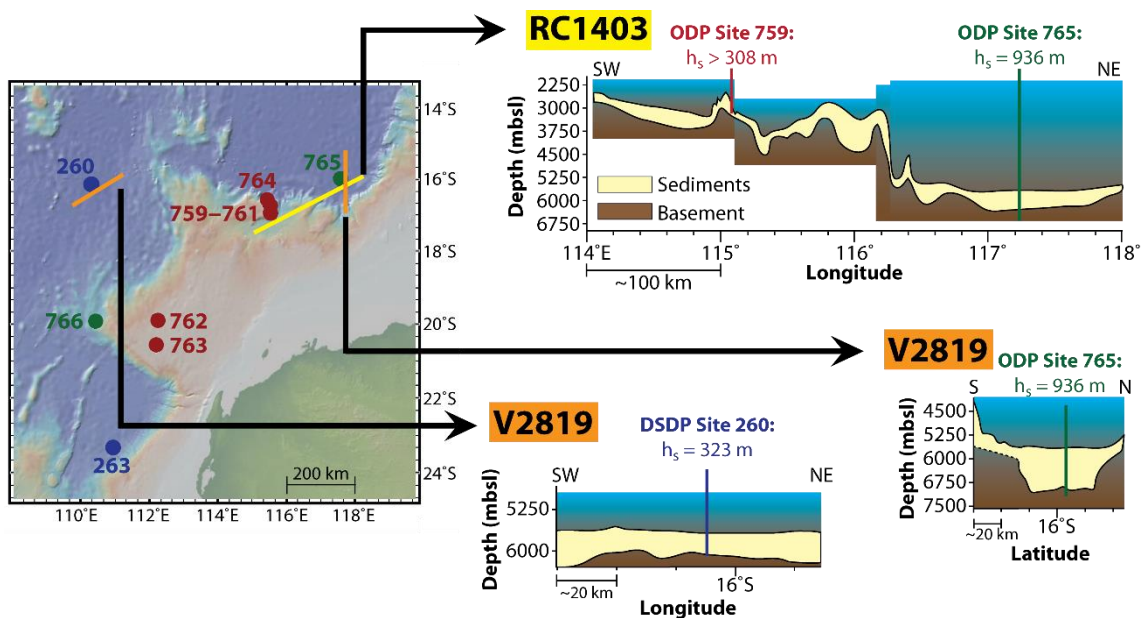


Figure 3.12: Single-Channel Seismic (SCS) reflection profiles RC1403 and ELT45, with correspondent locations on topographic map [modified from <http://www.geomapapp.org>]. The DSDP-ODP sites are highlighted with colours, and the corresponding measured sediment thickness is indicated as " $h_s$ ".

The equilibrium temperature at Site 765 (37 °C, Table 3.4) cannot be used in the validation of oceanic crustal age-derived temperature at the sediment-basement interface because it has been computed following the *Bullard method* [Bullard, 1947] based on a heat-conduction model, which in turn is at the base of the temperature estimations in this study. However, the temperature estimated with the GDH1 model at

Site 765, using  $K_s = 1 \text{ W/(m K)}$ , and interpolating values of sediment thickness, is  $\sim 37^\circ \text{C}$  (Table 3.4), in perfect agreement with the equilibrium value proposed by *Castillo* [1992] specifically for this location. The same temperature is calculated despite the  $\sim 200 \text{ m}$  difference in sediment thickness between model and measurements, due to higher thermal conductivity considered by *Castillo* [1992] ( $\sim 0.3 \text{ W/(m K)}$  higher than GDH1 model).

### 3.5.2.2. Implications of target a for offshore CCS

*Target a* has been defined on the basis of combined gravitational ( $\Delta\rho > 0 \text{ kg/m}^3$ ) and physical trapping (sediment thickness between 200 and 700 m) (Figure 3.11). Its areal extent is  $\sim 1.47 \times 10^6 \text{ km}^2$  (Table 3.2), but the water depth is on average greater than 5000 m, and a very limited part of the target is within 500 km of the Australian coastline. These characteristics make *target a* theoretically suitable for geological carbon sequestration, but practically incompatible with current technology and markets that are mostly interested in shallow reservoirs that are easy to monitor and have short transport distances [*Rubin*, 2008]. However, if only gravitational trapping is applied, *target a* almost doubles its size, including regions previously excluded because of sediment thicknesses below the 200 m threshold (Figure 3.13). In particular, not too far away from the southern coast of Western Australia, several square kilometres of oceanic crust become available (*target a1* in Figure 3.13), with sediment thickness between 100 and 200 m, but still located in deep waters ( $\sim 5000 \text{ m}$ ). At these conditions, the liquid  $\text{CO}_2$ , naturally denser than seawater, would have the tendency to sink within the basement, and be stable with only a thin or absent sediment cover.

The potential carbon sequestration in *target a*, by  $\text{CO}_2$  storage in deep-sea basalts based on gravitational trapping, is not the first project suggesting injections of  $\text{CO}_2$  in offshore Australia. The offshore region in NW Australia is an active region of natural gas production. For example, the Gorgon Project (Figure 3.13) is one of the largest natural

gas projects ever undertaken in the world. It is located ~200 km from the northwest coast of Western Australia, and includes the injection and storage of CO<sub>2</sub> into a deep sedimentary formation sited 2.3 km below Barrow Island [<https://www.chevronaustralia.com/our-businesses/gorgon>]. The gas extracted from the Gorgon Field is separated from the naturally occurring reservoir CO<sub>2</sub> (~14%) before being liquefied in a facility capable of producing 15.6 Mt of liquid natural gas (LNG) per year. Liquefying natural gas to -162 °C has the advantage of decreasing the volume of the gas 600 times, making shipping around the world more economical. The removed CO<sub>2</sub> will be transported via pipeline and injected beneath Barrow Island, migrating through the aquifer until it becomes trapped. According to preliminary estimations, this will reduce greenhouse gas emissions from the project by ~40% [<https://www.chevronaustralia.com/our-businesses/gorgon>].

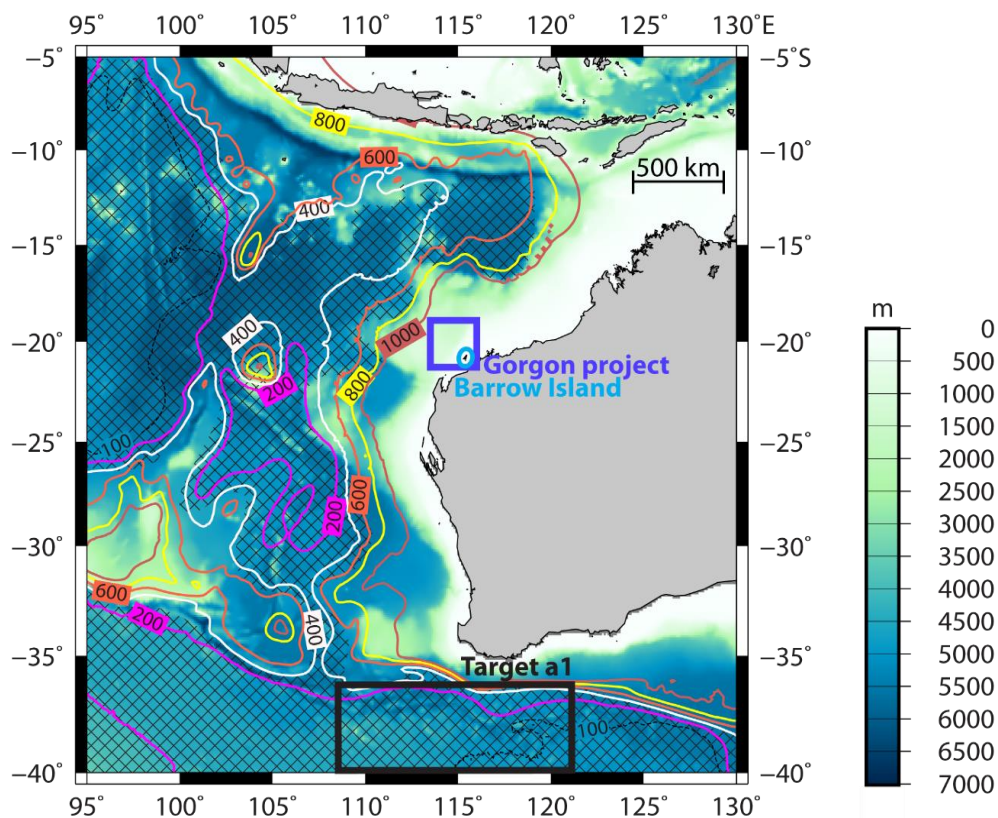


Figure 3.13: Bathymetric map showing the potential target **a** for CO<sub>2</sub> storage. Suitable locations for gravitational trapping ( $\Delta\rho > 0 \text{ kg/m}^3$ ) are hatched, and target **a1** is shown in a black box. Sediment thickness is represented by coloured lines every 200 m, whereas the 100 m isoline is shown as a black dashed line. The location of the Gorgon Project is highlighted on the map with a blue square, and the position of Barrow Island with a light blue circle.

Hence, considering both scenarios of CO<sub>2</sub> storage offshore, *target a* and the Gorgon Project, two main questions can be raised:

- 1) Is the shipping of liquefied gas used in the oil industry a viable options for CO<sub>2</sub> transport to offshore reservoirs?
- 2) Why has no CO<sub>2</sub> re-injection project been undertaken yet in oceanic crust, where the advantages derived from the presence of *in situ* gravitational trapping, a low-permeability sediment cap, and potentially highly reactive rock can be exploited?

An attempt to address these questions has been made in Chapter 5, analysing the CO<sub>2</sub> costs linked with CCS offshore projects.

### 3.6. Conclusions

The conceptual storage of carbon dioxide in mafic formations offshore is a mitigation strategy that is still developing.

In this chapter, limitations and constraints on its applicability have been given, considering gravitational and physical trapping in oceanic crust with the use of global databases. The results can be summarised as follows:

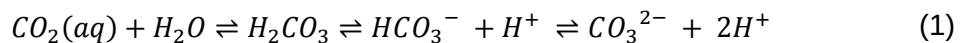
- the general carbon sequestration potential of oceanic crust must be evaluated starting from global scale studies on CO<sub>2</sub> stability at the sediment-basement interface, and proceeding towards more local analyses to better understand the relationships between observations at different scales;
- the extent of suitable areas for geological CO<sub>2</sub> sequestration depends on the level of physical and gravitational stability of CO<sub>2</sub> that is implemented in the evaluations;
- ~48% of the oceanic crust at the sediment-basement interface is suitable for gravitational trapping ( $\Delta\rho_{\text{CO}_2\text{-seawater}} > 0 \text{ kg/m}^3$ ), and ~32% for physical trapping (sediment thickness between 200 and 700 m);
- the combination of gravitational and physical trapping reduces the oceanic crust suitable for CCS to ~8%; using these criteria, five potential targets have been selected;
- even the smallest identified reservoir could provide sufficient carbon dioxide sequestration capacity for several centuries;
- sites on the Juan de Fuca Plate and in the eastern equatorial Pacific Ocean are unsuitable for gravitational trapping of carbon dioxide because the CO<sub>2</sub> will naturally tend to escape due to the thermal conditions at the sediment-basement interface;
- specific investigations on *target a* (Indian Ocean) show the limitation of global scale studies. This highlights the need for detailed programs of local data acquisition (e.g., sediment thickness, heat flow, temperature, thermal conductivity, porosity, and permeability) to further develop the CO<sub>2</sub> trapping potential of this area.

## Chapter 4: Low temperature batch dissolution experiments in a CO<sub>2</sub>-seawater-rock system

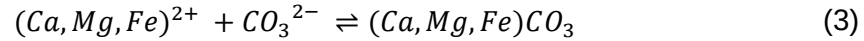
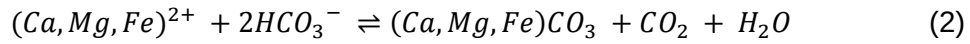
### 4.1. Introduction

Among the several mitigation strategies proposed to sequester CO<sub>2</sub>, *mineral carbonation* is one of the most promising [Lackner, 2003; Seifritz, 1990; Sipilä *et al.*, 2008]. As described in Chapter 3, mineral carbonation process is a *geochemical trapping* mechanism, where purified CO<sub>2</sub> and water react with the surrounding rocks to form geologically stable carbonate minerals [Lackner *et al.*, 1995]. This may reduce the likelihood of accidental release and the need for post-storage monitoring [Hawkins, 2004; Rochelle *et al.*, 2004]. This mechanism has been found to be particularly efficient (time required <1000 yr) in mafic and ultramafic formations [Galeczka *et al.*, 2014; Gislason *et al.*, 2010; Kelemen and Matter, 2008; Paukert *et al.*, 2012; Schaef *et al.*, 2010; Van Pham *et al.*, 2012; Xu *et al.*, 2005]. Rocks rich in calcium and magnesium silicate minerals, such as basalts, can convert CO<sub>2</sub> to carbonate minerals by providing divalent cations, such as Ca<sup>2+</sup>, Mg<sup>2+</sup> and Fe<sup>2+</sup> [Goldberg, 1999; Matter *et al.*, 2007].

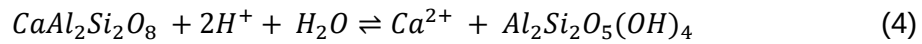
Several reactions are involved with geochemical trapping in basaltic rocks [Takahashi *et al.*, 2000]. Carbon dioxide dissolves into formation waters through the following acidification reactions:



The concentration of dissolved CO<sub>2</sub> in solution depends on CO<sub>2</sub> solubility, which in turn is a function of temperature, pressure and salinity of the formation water [Portier and Rochelle, 2005]. The products of CO<sub>2</sub> dissociation are bicarbonate (HCO<sub>3</sub><sup>-</sup>) and carbonate (CO<sub>3</sub><sup>2-</sup>) ions (Equation 1), which can precipitate as carbonate minerals if divalent cations are available in solution (Equations 2 and 3).



The production of  $H^+$  can be a limiting step in the precipitation of carbonates. Hence, further water-rock interactions are required to consume  $H^+$  ions and drive reactions (1) and (2) to the right [Matter and Kelemen, 2009]. The dissolution of calcium plagioclase – anorthite  $CaAl_2Si_2O_8$  – (Equation 4) is an example of a forward reaction that consumes hydrogen ions and produces  $Ca^{2+}$ :



Oelkers *et al.* [2008] investigated the potential of anorthite, as well as other silicate minerals and basaltic glass, as a Ca-source for mineral carbonation. The results, based on mineral stoichiometry, suggest that 23.1 t of Ca-plagioclase are required to sequester a tonne of carbon as calcium carbonate - calcite  $CaCO_3$ , whereas only 8.76 t of basaltic glass are needed to convert the same amount of carbon in a suite of carbonate minerals (calcite, magnesite  $MgCO_3$ , and siderite  $FeCO_3$ ).

Analyses of the upper oceanic crust show that interactions between seawater and basalt affect the global carbon budget, fixing dissolved  $CO_2$  as carbonate minerals in rocks and veins at low temperature ( $<60^\circ C$ ) [Alt and Teagle, 1999; Coogan and Gillis, 2013; Coogan *et al.*, 2016]. The net annual uptake rate of carbon as a result of oceanic crust alteration is  $\sim 1 \times 10^{12}$  mol C  $yr^{-1}$ , and varies with oceanic crustal age and location [Alt and Teagle, 1999; Gillis and Coogan, 2011; Staudigel *et al.*, 1989]. Due to prolonged and extensive faulting, slow-spreading Atlantic crust can store higher quantities of carbonate in young crust, whereas fast-spreading Pacific crust shows a steady increase in carbonate abundance over time [Rausch, 2012]. These studies indicate that  $>80\%$  of the carbonate minerals form during the hydrothermal alteration on the ridge flank within  $<25$  Myr of oceanic crust formation [Coogan *et al.*, 2016; Harris *et al.*, 2014]. This means that, although fast in geological terms, these geochemical reactions proceed very slowly



on human time scale. Hence, questions remain about whether the carbonation process can be speeded up, in order to exploit the oceanic crust for CO<sub>2</sub> storage in a relatively short period of time by mineral carbonation.

As shown by *Coogan and Gillis* [2013] comparing Mesozoic (higher atmospheric CO<sub>2</sub> concentration) and Cenozoic (lower atmospheric CO<sub>2</sub> concentration) silicate-carbonate weathering rates, the increase in alkalinity at high CO<sub>2</sub> concentration is due to the breakdown of igneous minerals and the consequent release of cations, which favours carbonate precipitation. In most cases though, silicate dissolution rates are slower than the corresponding carbonate precipitation rates, so accelerating the dissolution of silicate minerals is the key to optimize the mineralisation [*Oelkers et al.*, 2008].

Mineral dissolution rates in mafic and ultramafic rocks are enhanced by increasing the reactive surface area of the rock by grinding the material, by increasing temperature, and by adding acids and bases (such as H<sub>2</sub>SO<sub>4</sub>, HNO<sub>3</sub>, NaOH, and NaHCO<sub>3</sub>) [*Anbeek*, 1992; *Blum and Lasaga*, 1988; *Grandstaff*, 1978; *O'Connor et al.*, 2000; *Oelkers*, 2001a; *Teir et al.*, 2007]. Over the last two decades, a number of experiments have provided better understanding of the chemical processes controlling the dissolution of mafic and ultramafic rocks in CO<sub>2</sub>-rich solutions, and in far-from equilibrium solutions. Particularly, experimental approaches include studies on crystalline basalt and basaltic glass at low temperature (25 – 75 °C) [*Galeczka et al.*, 2014; *Gislason and Oelkers*, 2003; *Gudbrandsson et al.*, 2011; *Gysi and Stefánsson*, 2012c; *Oelkers and Gislason*, 2001], and at high temperature (75 -250 °C) [*Gysi and Stefánsson*, 2012a; b]; on metabasalt [*Critelli et al.*, 2014]; on peridotite [*Andreani et al.*, 2009]; and on both basalt and peridotite in the presence of seawater [*Wolff-Boenisch et al.*, 2011]. More information on these experiments are provided in Table 4.1. In general, based on measurements of silicon release rates, it has been observed that under specific conditions (e.g., pH, pCO<sub>2</sub>, and temperature) basalt dissolution rates may be nearly as high as those for peridotite [*Matter and Kelemen*, 2009; *Wolff-Boenisch et al.*, 2011] (Table 4.2).

Table 4.1: Summary of the experiments on mafic and ultramafic rocks carried out in the last two decades to investigate the rock dissolution potential.

|                                      | Reactor setup             | Rock                                           | Initial conditions |                         |         |                                                                                                  |
|--------------------------------------|---------------------------|------------------------------------------------|--------------------|-------------------------|---------|--------------------------------------------------------------------------------------------------|
|                                      |                           |                                                | T<br>°C            | pCO <sub>2</sub><br>bar | pH      | Solution                                                                                         |
| <i>Andreani et al. [2009]</i>        | <sup>a</sup> flow-through | sintered peridotite                            | 160                | 110                     | 6.7     | H <sub>2</sub> O + HCl + CO <sub>2</sub> + NaHCO <sub>3</sub>                                    |
| <i>Critelli et al. [2014]</i>        | <sup>b</sup> mixed flow   | metabasalt                                     | 25                 | -                       | 2-12    | H <sub>2</sub> O + HCl + NaCl + NaOH                                                             |
| <i>Galeczka et al. [2014]</i>        | <sup>a</sup> flow-through | basaltic glass                                 | 22, 50             | 10 <sup>-5.7</sup> , 22 | 6.7     | H <sub>2</sub> O + CO <sub>2</sub>                                                               |
| <i>Gislason and Oelkers [2003]</i>   | <sup>b</sup> mixed flow   | basaltic glass                                 | 6-150              | -                       | 2-11    | H <sub>2</sub> O ± HCl ± NH <sub>4</sub> Cl ± NH <sub>3</sub>                                    |
| <i>Gudbrandsson et al. [2011]</i>    | <sup>b</sup> mixed flow   | crystalline basalt                             | 5-75               | -                       | 2-11    | H <sub>2</sub> O ± HCl ± NH <sub>4</sub> OH ± NH <sub>4</sub> Cl                                 |
| <i>Gysi and Stefansson [2012]a,c</i> | <sup>c</sup> mixed batch  | basaltic glass                                 | 75-250             | 11-24                   | 4.5-5.9 | Vellankatla spring + CO <sub>2</sub>                                                             |
| <i>Gysi and Stefansson [2012]b</i>   | <sup>c</sup> mixed batch  | basaltic glass                                 | 40                 | 1-13                    | 3.6-4.5 | Vellankatla spring + CO <sub>2</sub>                                                             |
| <i>Oelkers and Gislason [2001]</i>   | <sup>b</sup> mixed flow   | basaltic glass                                 | 25                 | -                       | 3-11    | H <sub>2</sub> O + HCl + NaOH + C <sub>2</sub> H <sub>2</sub> O <sub>4</sub> + AlCl <sub>3</sub> |
| <i>Wolff-Boenisch et al. [2011]</i>  | <sup>b</sup> mixed flow   | basaltic glass, crystalline basalt, peridotite | 25                 | 4                       | 3.6     | Seawater + CO <sub>2</sub>                                                                       |

<sup>a</sup>flow-through: the solution passes through the reactor, and gets progressively discharged;

<sup>b</sup>mixed flow: the solution flows into the reactor, is mixed with a stirrer, and leaves the reactor;

<sup>c</sup>mixed batch: the solution is present from the beginning in the reactor, and is mixed with a stirrer.

Table 4.2: Summary of previous experimental results on rock dissolution potential, described as maximum silicon release rates, normalised to surface area. Experimental conditions are also shown.

|                                     | Rock               | T<br>°C | pCO <sub>2</sub><br>bar | pH    | Surface area analysis | Si release rate<br>log (mol/cm <sup>2</sup> /s) |
|-------------------------------------|--------------------|---------|-------------------------|-------|-----------------------|-------------------------------------------------|
| <i>Critelli et al. [2014]</i>       | metabasalt         | 25      | -                       | 2-12  | BET                   | from -14.3 to -15.9                             |
| <i>Galeczka et al. [2014]</i>       | basaltic glass     | 50      | 22                      | 4.4   | geo                   | -13.2                                           |
| <i>Gislason and Oelkers [2003]</i>  | basaltic glass     | 50      | -                       | 2     | BET, geo              | -10                                             |
| <i>Gudbrandsson et al. [2011]</i>   | crystalline basalt | 75      | -                       | 10.96 | BET                   | -13.3                                           |
| <i>Oelkers and Gislason [2001]</i>  | basaltic glass     | 25      | -                       | 3     | BET                   | -12.5                                           |
| <i>Wolff-Boenisch et al. [2011]</i> | basaltic glass     | 25      | 4                       | 3.76  | geo                   | -12.42                                          |
|                                     | crystalline basalt | 25      | 4                       | 3.76  | geo                   | -12.14                                          |
|                                     | peridotite         | 25      | 4                       | 3.76  | geo                   | -11.65                                          |

However, many of the uncertainties related to the CO<sub>2</sub>-water-rock interactions are still under investigation; for example:

- Which mafic to ultramafic rocks provide the best physical and geochemical conditions for CO<sub>2</sub> sequestration?
- Is there any difference in reactivity between offshore and onshore basalts, and ophiolitic rocks?
- How complete and comparable are dissolution rate results obtained from different experiments?
- What are the implications related to the use of complex crystalline rocks during CO<sub>2</sub>-experiments instead of mono-mineral or homogeneous (basaltic glass) materials?

In this chapter, primary rock dissolution rates and key reactions associated with CO<sub>2</sub> geochemical trapping in mafic rocks are investigated through batch dissolution experiments in a CO<sub>2</sub>-seawater-rock system at CO<sub>2</sub> partial pressure (pCO<sub>2</sub>) of ~1 bar and 40 °C. This temperature has been chosen because it is typical of ridge flank conditions for low temperature hydrothermal alteration, and comparable with other studies conducted on basalt in presence of seawater [Crovisier *et al.*, 1987; Seyfried and Mottl, 1982; Wolff-Boenisch *et al.*, 2011]. A major difference between this study and previous studies of CO<sub>2</sub>-water-rock interactions is the nature of the experimental material. For the first time, rocks from the upper oceanic crust and ophiolitic rocks have been used in CO<sub>2</sub>-mineral dissolution experiments.

## **4.2. Material and methods**

### **4.2.1. Starting material**

#### Reactive fluid

The seawater used in the experiments (CaribSea Inc) has a salinity of ~35 psu, and the composition is in agreement with IAPSO (International Association for the Physical Sciences of the Oceans) seawater [Millero *et al.*, 2008; Summerhayes and Thorpe, 1996]

(Table 4.3). It was stored at 4 °C in the dark and warmed up to room temperature overnight prior to its use in the experiments.

pH was measured with a semi micro plastic BNC electrode (Fisherbrand™), connected to a pH meter (Accumet AB 15/15+, Fisher Scientific™). The electrode accuracy was calibrated against standard buffer solutions, with a standard deviation of 0.03 pH units.

Total Alkalinity (TA) was determined with the Gran function using a modified titration methodology for a small-volume system (Appendix “Methods” CH4-1) [Haraldsson *et al.*, 1997].

Dissolved Inorganic Carbon (DIC) was calculated with the following equation:

$$DIC = \frac{Ac([H^+]^2/K_1 + [H^+] + K_2)}{[H^+] + 2K_2} \quad (5)$$

with  $[H^+]$  derived from pH, Ac (carbon Alkalinity) assumed equal to TA, and the dissociation constants  $K_1 = 1.89 \times 10^{-6}$  and  $K_2 = 1.85 \times 10^{-9}$  estimated for  $T = 40\text{ °C}$ ,  $pCO_2 \sim 1$  bar, and  $S = 35$  psu, using the database available in *Millero et al.* [2006].

The Si, Na, Mg, Ca, K, and S concentrations in seawater were measured by inductively coupled plasma optical emission spectrometry (ICP–OES, Perkin-Elmer Optima 4300 DV). Fe was not measured due to potential issues with contamination from the stainless steel bubbler (see Section 4.2.3). The reproducibility of the ICP–OES analyses, determined by replicate analysis of the same sample is better than  $\pm 3\%$  for all elements. A detailed explanation on the preparation of samples and standards for ICP-OES analysis can be found in Appendix “Methods” CH4-2. Concentrations of  $SO_4$  and Cl were measured by ion chromatography (IC, Dionex ICS2500). Repeat analysis of IAPSO seawater as well as single anion standards indicates that the reproducibility of the Cl and sulphate analyses is better than  $\pm 3\%$ . A detailed explanation on the preparation of samples and standards for IC analysis can be found in Appendix “Methods” CH4-3.

Table 4.3: Chemical composition of the starting solution for the experimental work. Values are given in mmol/L.

|                             | pH / °C   | TA   | DIC  | Si   | Na  | Mg   | Ca   | K    | S    | SO <sub>4</sub> | Cl  |
|-----------------------------|-----------|------|------|------|-----|------|------|------|------|-----------------|-----|
| <b>IAPSO</b><br>reference*  | 8.10 / 20 | 2.30 | 1.94 | 0.10 | 469 | 53.1 | 10.3 | 10.2 | 28.2 | 28.2            | 546 |
| <b>IAPSO</b><br>measured    | 7.66 / 20 | 2.38 | 2.23 | 0.11 | 468 | 52.6 | 10.5 | 10.0 | 28.1 | 27.5            | 555 |
| <b>SEAWATER</b><br>measured | 7.46 / 20 | 2.43 | 2.36 | 0.02 | 469 | 50.6 | 10.5 | 10.1 | 28.7 | 28.5            | 565 |

\* from Summerhayes and Thorpe [1996], and Millero et al. [2008]

## Rocks

The rocks used in this study are mid-ocean ridge basalts from the Juan de Fuca and Mid-Atlantic Ridges, and a gabbro from the Troodos ophiolite in Cyprus. The mineralogy was determined by optical microscopy of polished thin sections (data summarised in Table 4.4), and confirmed by scanning electron microscope (SEM) analyses (Leo 1450VP SEM combined with Oxford Instruments X-Act 10mm<sup>2</sup> area SDD EDS Detector). Also, elemental line scans and mineral phase mapping were determined using the AZtec Energy software package connected to the SEM instrumentation. Percentages of phase concentrations are relative to the mapped area, which is ~10% of the whole thin section, and do not always reflect the actual volume of mineral in thin section, especially with cryptocrystalline texture. On the other hand, the maps provide SEM-EDS analyses of each phase present in the rock. Full SEM-EDS analysis results can be found in Appendix “Analyses” CH4-6.

The **Juan de Fuca** (JdF) samples consist of forty crystalline basalts collected during the scientific cruise *IODP Exp 327* from the *Hole U1362A*. They are heterogeneous multi-mineral solids that consist on average of 38 vol% labradoritic plagioclase, 23 vol% augitic clinopyroxene, 14 vol% mesostasis, 15 vol% secondary minerals, with minor olivine, iron oxides, and glass. The secondary mineralogy is composed of saponite, celadonite, oxides, and, if present, ~1 vol% calcite (Table 4.4). This suite is typical of low temperature hydrothermal alteration, and is mainly associated with groundmass replacement, vesicle

fill, and alteration halos [Fisher *et al.*, 2011]. Due to the small volume of individual specimen (~12 cm<sup>3</sup> each), they were combined together to form three composite samples, on the basis of textural differences and composition. These composite samples, corresponding to cryptocrystalline, microcrystalline, and fine grained basalts, are identified as “JdF mix 1”, “JdF mix 2”, and “JdF mix 3”, respectively. The complete list of samples and corresponding images, divided by textural group, is available in Appendix “Samples” CH4-13.

Table 4.4: Mineralogical composition of the MOR basalts (JdF mix 1, 2, 3, and CD80WP132) and ophiolitic gabbro (G1). The terminology definitions can be found in Appendix “Methods” CH4-4.

|                               | JDF MIX 1                                  | JDF MIX 2                                   | JDF MIX 3                                         | CD80WP132                                | G1                         |
|-------------------------------|--------------------------------------------|---------------------------------------------|---------------------------------------------------|------------------------------------------|----------------------------|
| <b>SAMPLING METHOD</b>        | drilling                                   | drilling                                    | drilling                                          | dredging                                 | fieldwork                  |
| <b>LITHOLOGY</b>              | MOR basalt                                 | MOR basalt                                  | MOR basalt                                        | MOR basalt                               | ophiolitic gabbro          |
| <b>GRAIN SIZE</b>             | cryptocrystalline <sup>a</sup>             | microcrystalline <sup>b</sup>               | microcrystalline <sup>b</sup> / fine <sup>c</sup> | cryptocrystalline <sup>a</sup>           | medium <sup>d</sup>        |
| <b>TEXTURE</b>                | Intersertal glomeroporphyritic cvariolitic | Intergranular glomeroporphyritic variolitic | Intergranular subophitic                          | Intersertal glomeroporphyritic vesicular | intergranular granoblastic |
| <b>PRIMARY MINERALOGY %</b>   | <b>84</b>                                  | <b>80</b>                                   | <b>78</b>                                         | <b>85</b>                                | <b>42</b>                  |
| <b><u>PHENOCRYSTS %</u></b>   | <b>19</b>                                  | <b>15</b>                                   | <b>6</b>                                          | <b>35</b>                                | <b>-</b>                   |
| Plagioclase-labradorite       | 14                                         | 11                                          | 4                                                 | 20                                       | -                          |
| Pyroxene-augite               | 4                                          | 4                                           | 2                                                 | 12                                       | -                          |
| Olivine                       | 1                                          | trace                                       | <1                                                | 3                                        | -                          |
| <b><u>GROUNDMASS %</u></b>    | <b>65</b>                                  | <b>65</b>                                   | <b>72</b>                                         | <b>50</b>                                | <b>42</b>                  |
| Plagioclase-labradorite       | 28                                         | 30                                          | 28                                                | 11                                       | -                          |
| Plagioclase-anorthite         | -                                          | -                                           | -                                                 | -                                        | 40                         |
| Pyroxene-augite               | 19                                         | 20                                          | 20                                                | 5                                        | 2                          |
| Mesostasis                    | 16                                         | 10                                          | 17                                                | 11                                       | -                          |
| Opakes                        | 2                                          | 5                                           | 5                                                 | -                                        | -                          |
| Olivine                       | -                                          | trace                                       | -                                                 | 3                                        | -                          |
| Glass                         | -                                          | -                                           | 2                                                 | 20                                       | -                          |
| <b>SECONDARY MINERALOGY %</b> | <b>13</b>                                  | <b>13</b>                                   | <b>19</b>                                         | <b>&lt;1</b>                             | <b>58</b>                  |
| Saponite                      | 8                                          | 8                                           | 10                                                | -                                        | -                          |
| Celadonite                    | 2                                          | 1                                           | 6                                                 | -                                        | -                          |
| Oxides                        | 3                                          | 3                                           | 3                                                 | <1                                       | -                          |
| Calcite                       | 1                                          | 1                                           | -                                                 | -                                        | -                          |
| Amphibole-actinolite          | -                                          | -                                           | -                                                 | -                                        | 40                         |
| Plagioclase-albite            | -                                          | -                                           | -                                                 | -                                        | 13                         |
| Chlorite                      | -                                          | -                                           | -                                                 | -                                        | 3                          |
| Talc                          | -                                          | -                                           | -                                                 | -                                        | 2                          |
| <b>VESICLES MINERALOGY %</b>  | <b>4</b>                                   | <b>7</b>                                    | <b>3</b>                                          | <b>15</b>                                | <b>-</b>                   |
| Minerals                      | celadonite, saponite, oxides               | celadonite, saponite, oxides                | celadonite, saponite, oxides                      | -                                        | -                          |
| Filled                        | 70%                                        | 60%                                         | 60%                                               | <1%                                      | -                          |

<sup>a</sup> cryptocrystalline: crystals <0.1 mm; <sup>b</sup> microcrystalline: 0.1 mm < crystals < 0.2 mm; <sup>c</sup> fine grained: 0.2 mm < crystals < 1 mm; <sup>d</sup> medium grained: 1 mm < crystals < 5 mm.

The pillow basalt from the **Mid-Atlantic Ridge** (MAR) was dredged from the Reykjanes Ridge, SW of Iceland, during the *RRS Charles Darwin Cruise 80*, specifically at the Way

*Point 132* (28.66°W, 60.54°N), and was provided for this study by Dr. Bramley J. Murton (*National Oceanography Centre, Southampton*). Mineralogically, the sample is a cryptocrystalline vesicular basalt consisting of 31 vol% labradoritic plagioclase, 20 vol% glass, 17 vol% augitic clinopyroxene, 11 vol% mesostasis, 6 vol% olivine, <1 vol% secondary minerals, and 15 vol% unfilled vesicles (Table 4.4). From now on, the MAR basalt is called “CD80132WP” (image available in Appendix “*Samples*” CH4-14).

The gabbro from the **Troodos ophiolite** (Cyprus) was sampled by Prof. Damon A.H. Teagle during a fieldwork in May 2014, close to the village of Kato Amiandos (32.93°E, 34.93°N). The rock shows a high degree of hydrothermal alteration, with mineralogy consisting of 40 vol% anorthitic plagioclase, 40 vol% actinolitic amphibole (replacing clinopyroxene), 13 vol% albitic plagioclase, 3 vol% chlorite, 2 vol% augitic clinopyroxene, and 2 vol% talc (Table 4.4). From now on, the Troodos gabbro will be called “G1” (image available in Appendix “*Samples*” CH4-15).

All solids were ground to obtain the 63 – 125 µm size fraction (Section 4.2.2.). The composition of 63-125 µm and <63 µm size fractions, were determined for each rock by inductively coupled plasma mass spectrometry (ICP-MS, Thermo Scientific X-Series 2), and X-ray diffraction (XRD) (Appendices “*Analyses*” CH4-7, and -8, respectively). A detailed description of the methodology used for ICP-MS analysis is available in Appendix “*Methods*” CH4-5. Furthermore, chemical analyses with X-ray fluorescence spectroscopy (XRF) were run at the University of St. Andrews (Scotland, UK), and are here summarised in Table 4.5 for 63-125 µm and <63 µm size fractions. The complete list of XRF analyses, with major and trace elements for both 63-125 µm and <63 µm size fractions, is available in Appendix “*Analyses*” CH4-9. The rock compositions in this study are considered from the XRF analyses, which show better accuracy and precision than ICP-MS analyses (Appendix “*Analyses*” CH4-9, and “*Methods*” CH4-5, respectively).

In general, rock compositions are similar to that of mid ocean ridge basalt – MORB – summarised by *GERM* [2000], as confirmed by ICP-MS analyses. No major differences are found between the two size fractions. Loss on ignition (LOI) is the percentage weight lost on sample ignition, due to the loss of volatile phases (for example H<sub>2</sub>O, CO<sub>2</sub> and S) that may in part be compensated by a gain in mass due to oxidation of Fe<sup>2+</sup> to Fe<sup>3+</sup>. However, the dominant control on LOI is the concentration of water, which increases as a result of hydration of primary minerals during alteration. This effect is superimposed on the primary water content of the rocks compared to typical MORB (<0.2 wt% or less) [Danyushevsky, 2001]. In general, the LOI is <3% for all the samples.

Table 4.5: Summary of rock chemical compositions for 63-125  $\mu\text{m}$  and <63  $\mu\text{m}$  size fractions, expressed in percent (data from XRF analyses). Negative values are most probably due to gain in mass due to oxidation of Fe<sup>2+</sup> to Fe<sup>3+</sup>.

| ID<br>%                        | JdF mix 1            |                   | JdF mix 2            |                   | JdF mix 3            |                   | CD80WP132            |                   | G1                   |                   |
|--------------------------------|----------------------|-------------------|----------------------|-------------------|----------------------|-------------------|----------------------|-------------------|----------------------|-------------------|
|                                | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ |
| SiO <sub>2</sub>               | 49.6                 | 48.3              | 48.6                 | 49.6              | 50.1                 | 49.6              | 49.0                 | 50.9              | 44.9                 | 47.4              |
| TiO <sub>2</sub>               | 1.93                 | 2.03              | 1.66                 | 1.60              | 1.70                 | 1.95              | 1.24                 | 1.05              | 0.0928               | 0.103             |
| Al <sub>2</sub> O <sub>3</sub> | 13.6                 | 13.3              | 13.7                 | 14.3              | 13.7                 | 14.2              | 12.5                 | 13.7              | 17.2                 | 14.4              |
| Fe <sub>2</sub> O <sub>3</sub> | 11.7                 | 12.4              | 11.2                 | 10.6              | 11.6                 | 12.2              | 14.5                 | 12.6              | 5.54                 | 6.44              |
| MnO                            | 0.233                | 0.237             | 0.200                | 0.190             | 0.200                | 0.208             | 0.222                | 0.190             | 0.109                | 0.126             |
| MgO                            | 6.47                 | 6.82              | 7.40                 | 7.18              | 6.79                 | 6.35              | 7.65                 | 7.57              | 11.1                 | 13.2              |
| CaO                            | 12.4                 | 12.9              | 13.9                 | 13.4              | 12.9                 | 12.8              | 12.9                 | 12.3              | 17.1                 | 14.1              |
| Na <sub>2</sub> O              | 2.35                 | 2.20              | 2.15                 | 2.09              | 2.12                 | 2.05              | 1.70                 | 1.51              | 0.949                | 1.08              |
| K <sub>2</sub> O               | 0.204                | 0.210             | 0.117                | 0.111             | 0.179                | 0.135             | 0.0248               | 0.0238            | 0.0426               | 0.0517            |
| P <sub>2</sub> O <sub>5</sub>  | 0.149                | 0.118             | 0.0739               | 0.0493            | 0.0589               | 0.0555            | 0.0539               | 0.0597            | <0.0110              | <0.0110           |
| SO <sub>3</sub>                | 0.102                | 0.0977            | 0.133                | 0.135             | 0.0780               | 0.110             | 0.244                | 0.212             | <0.0300              | <0.0300           |
| LOI                            | 1.33                 | 1.33              | 0.930                | 0.690             | 0.580                | 0.350             | -0.180               | -0.380            | 2.70                 | 2.90              |
| TOT                            | 100                  | 100               | 100                  | 100               | 100                  | 100               | 99.8                 | 99.6              | 99.7                 | 99.6              |

#### 4.2.2. Sample preparation

Most studies on mineral dissolution and carbon sequestration have been performed on finely ground samples, in order to achieve measurable results within a laboratory-scale time frame [Galeczka et al., 2014; Harrison et al., 2012; Oelkers and Gislason, 2001; Rosenbauer et al., 2012; Shibuya et al., 2013]. Gadikota et al. [2014b] described in detail a sample preparation protocol, highlighting the importance of accurate particle size characterisation, and the implications of smaller particles in the samples. The presence



of very fine particles increases the reactive surface area and extent of dissolution, giving false mineral reaction rates that cannot be compared to literature values. Hence, the current study attempted to utilise freshly ground rocks, which were ground following a multistep cleaning protocol, and provides a detailed physical description of the particles.

All the samples were ultrasonically cleaned and dried overnight at 60 °C, before being crushed with a laboratory ring mill (Rocklabs®) to reach a *fine sand* size fraction. To constrain the particle size between 63 and 125 µm and remove most of the fine particles produced during the grinding phase, a modified version of the cleaning procedure proposed by Gadikota *et al.* [2014b] was followed. The ground material was sieved using two stainless steel sieves of 63 and 125 µm, respectively (*dry sieving 1*), and cleaned in batches of 25 g. The 63-125 µm subsample obtained was then placed in the 63 µm sieve, previously positioned within a beaker, covered by deionized (DI) water, and cleaned ultrasonically for 5 minutes. At the end of the ultrasonication, the fine particles in suspension were removed and the sample still contained in the 63 µm sieve was flushed with DI water (*wet sieving*). This cycle of “ultrasonic bath-wet sieving” was repeated until no fine particles were observed. The final 63-125 µm sample was dried on a flat glass dish at 60°C overnight in a vacuum oven, to eliminate reaction between the cleaned minerals and air. A second dry sieving step at 63 µm was carried out (*dry sieving 2*) to ensure that no fine particles were still trapped in the sample. Once declared “fine particle-free”, the sample was weighed and stored in an amber glass bottle, which, in turn, was placed within a plastic bag to avoid any oxidation due to air contact.

The material loss for this type of cleaning procedure is quite high (~70%), because ~300 g of initial sample are needed to obtain ~80 g with a grain size fraction between 63 and 125 µm. This is a consideration to take into account in the selection of the starting material. Among all the samples used in this investigation, the ones from dredging (CD80WP132) and fieldwork (G1) were relatively easier to collect in big quantities (>1 kg) and no problem due to the loss of material emerged during the preparation. On the

other hand, every specimen from oceanic crust drilling (JdF) was at least two orders of magnitude smaller (~30 g each) than the other rocks, resulting in a significant sample loss. Also, in the powdered oceanic core samples, a higher abundance of fine particles (<10 µm) was observed, most probably due to the presence of secondary minerals, such as clays, which have a lower hardness than basaltic glass. Hence, the JdF specimens were prepared taking into account the composition, and acknowledging the presence of fine particles.

In light of these technical difficulties, the ground material (63-125 µm) was subjected to multiple physical analyses, specifically to constrain the grain size and morphology.

The Brunauer-Emmett-Teller (BET) multipoint technique (Quantachrome NovaWin BET Analyzer via nitrogen, Columbia University in the City of New York, USA) was used to determine the surface areas of the rock samples. Resulting BET surface areas ( $A_{\text{BET}}$ ) are listed in Table 4.6, together with the corresponding specific geometric surface areas ( $A_{\text{geo}}$ ) that were calculated assuming the grains are spherical (Equation 6):

$$A_{\text{geo}} = \frac{6}{(d \times \rho)} \quad (6)$$

where  $A$  is the total area (cm<sup>2</sup>/g),  $d$  the average particle diameter (cm), and  $\rho$  the density (g/cm<sup>3</sup>) [Cubillas *et al.*, 2005; Gautier *et al.*, 2001; Guy and Schott, 1988; Wolff-Boenisch *et al.*, 2004].

Table 4.6: Surface areas measured with BET ( $A_{\text{BET}}$ ) and geometrically calculated ( $A_{\text{geo}}$ ) for each of the five samples. Volumetric weighted means used as particle diameter averages in the  $A_{\text{geo}}$  calculations are shown. Density is assumed to be 2.9 g/cm<sup>3</sup> for all the samples.

|                  | $A_{\text{BET}}$     | $A_{\text{geo}}$     | Volumetric weighted<br>mean of particles |
|------------------|----------------------|----------------------|------------------------------------------|
|                  | [cm <sup>2</sup> /g] | [cm <sup>2</sup> /g] | [µm]                                     |
| <b>JdF mix 1</b> | 57240                | 406                  | 51                                       |
| <b>JdF mix 2</b> | 123420               | 489                  | 42                                       |
| <b>JdF mix 3</b> | 8760                 | 781                  | 27                                       |
| <b>CD80WP132</b> | 22140                | 188                  | 110                                      |
| <b>G1</b>        | 12670                | 191                  | 108                                      |

In a theoretical sample, ground to the size fraction 63-125  $\mu\text{m}$ , with 94  $\mu\text{m}$  as particle diameter mean and density of 2.9  $\text{g}/\text{cm}^3$ , the total geometrical surface area is 220  $\text{cm}^2/\text{g}$ . However, the abundance of clay and silt sized particles in the sample affects the average particle diameter and in turn the  $A_{\text{geo}}$ , which has to be defined sample by sample through more detailed analyses. Hence, grain size distribution (GSD) and relative particle diameter average were measured for each of the five samples using a Malvern Mastersize analyser (University of Southampton, UK) after shaking the samples overnight in a 0.05% Calgon solution to disaggregate the grains. GSD results are shown in Figure 4.1, whereas the volumetric weighted means used for  $A_{\text{geo}}$  calculations are reported in Table 4.6. Polished thin sections (PTS), and stubs were prepared for all the ground samples (63-125  $\mu\text{m}$ ) to further investigate and confirm the particle size distribution, composition and morphology by scanning electron microscopy (Leo 1450VP SEM combined with Oxford Instruments X-Act 10 $\text{mm}^2$  area SDD EDS Detector) (Figure 4.2). Complete SEM analyses on ground samples, including EDS and elemental mapping data, are available in Appendices “Analyses” CH4-10, and -11, respectively.

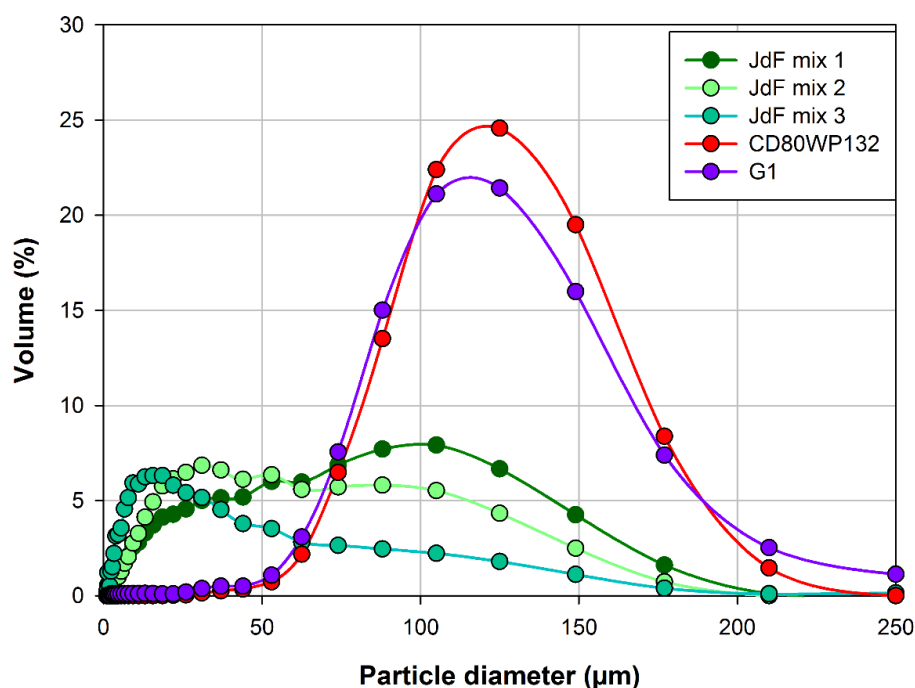


Figure 4.1: Grain size distribution (GSD) for all the ground samples (63-125  $\mu\text{m}$ ) before the dissolution experiments.

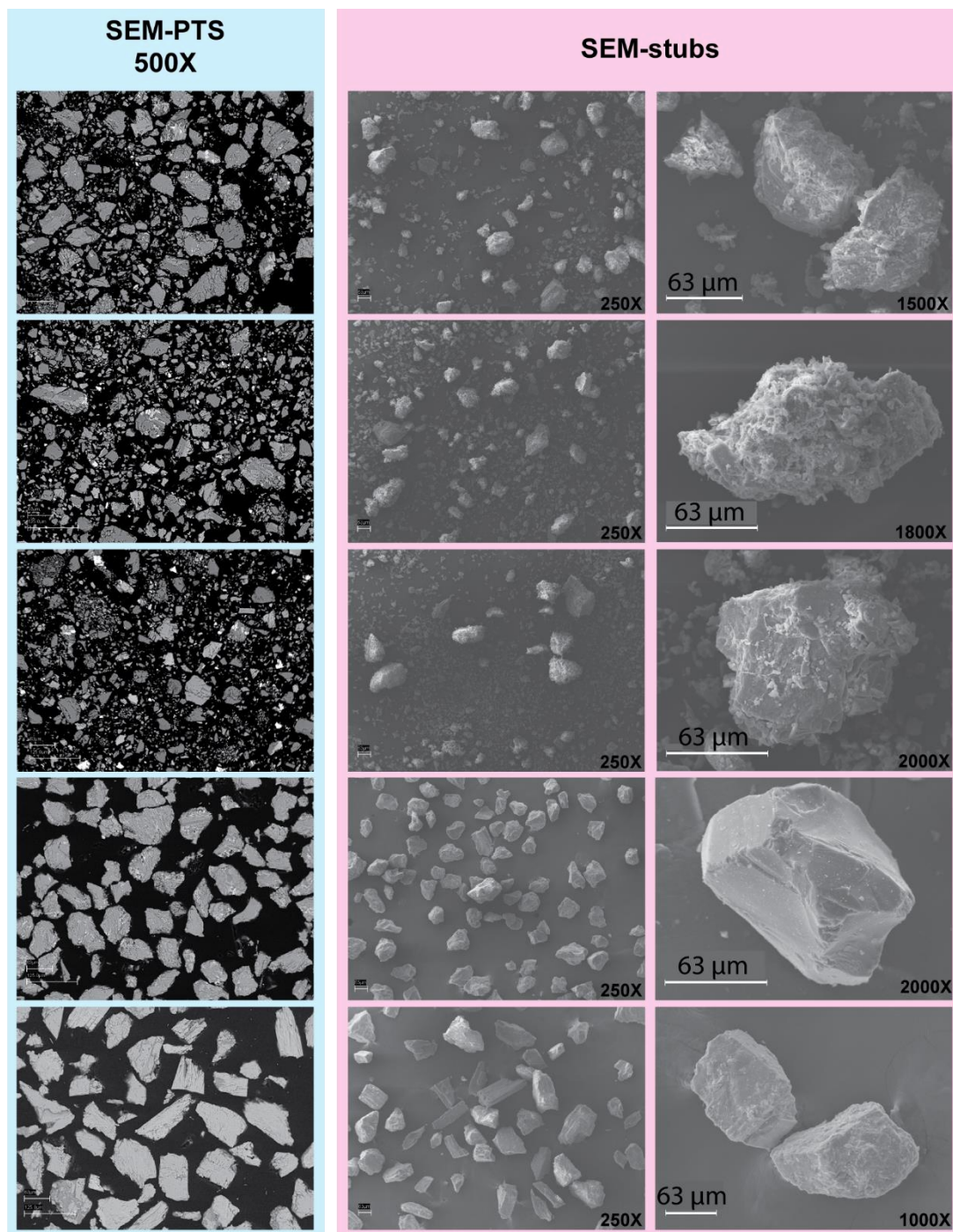


Figure 4.2: SEM images from PTS-polished thin sections (on the left, in the light blue box), and from stubs (on the right, in the pink box). The magnification for the PTS images is of 500X for all the samples, whereas the magnification for the stubs is indicated on each image.

#### 4.2.3. Experimental setup

The same experimental design was adopted in eight batch type experiments to determine the dissolution reaction rate of the three different rocks: JdF, CD80WP132, and G1. Based on the solid mass used in each experiments, three subseries can be identified: 80, 65, and 30 g. A schematic representation of all the performed experiments is shown in (Figure 4.3). The subseries of 80 g represents the standard initial rock mass following the method proposed by *Gysi and Stefánsson* [2012c] for moderate CO<sub>2</sub> concentrations, and involves five experiments: JdF mix 1, JdF mix 2, JdF mix 3, CD80WP132-A (a subsample of CD80WP132), and G1-A (a subsample of G1). The subseries of 65 g (JdF mix 2bis, CD80WP132-Abis) and 30 g (CD80WP132-B, subsample of CD80WP132) were run to better quantify the role of reactive surface area and fine particles during dissolution. Both JdF mix 2bis and CD80WP132-Abis experiments were performed on samples that had been previously used in the 80 g experiments (JdF mix 2 and CD80WP132-A, respectively), to further investigate the reactivity of these rocks in progressive dissolution experiments.

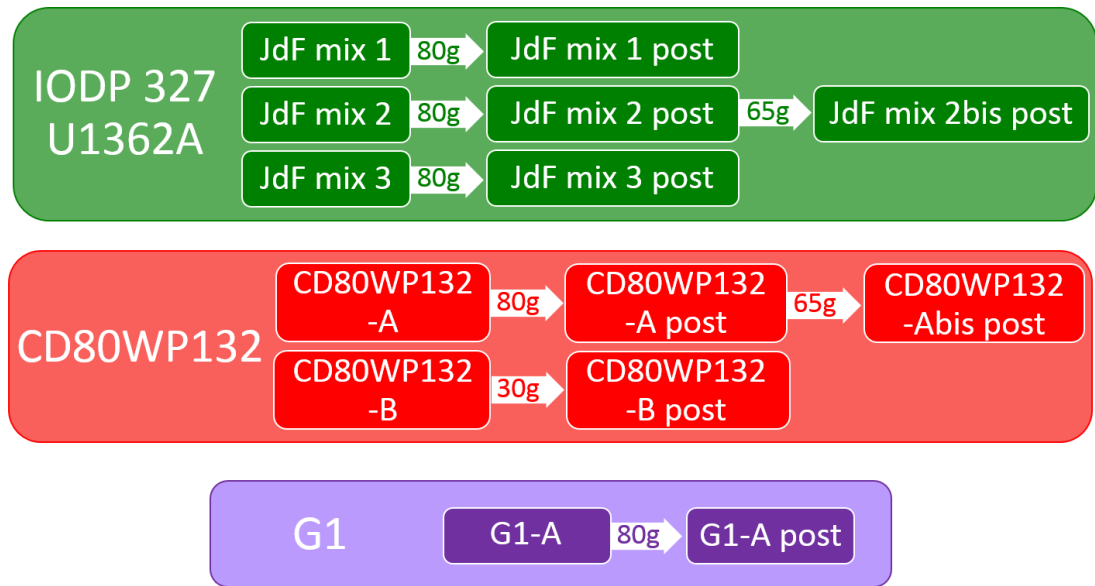


Figure 4.3: Summary of eight batch type experiments on three type of rocks: JdF (in green), CD80WP132 (in red), and G1 (in purple).

Several runs, lasting in total from 11 to 24 days, were carried out at 40 °C and  $p\text{CO}_2 \sim 1$  bar, in a three-phase system, consisting of oceanic rocks, seawater and gaseous  $\text{CO}_2$  in the headspace. The overall design is shown in Figure 4.4. Table 4.7 provides further details about the initial settings of each experiment.

For each run, 4 fluorinated HDPE 1000 mL bottles were positioned in a 25 L stainless steel water bath (MEDLINE BS-11). Each bottle was sealed with a Ley Rubber Ltd RB049/1H stopper with 4 holes: two of them for the inward and outward  $\text{CO}_2$  flux, respectively; one for the pH electrode (Fisherbrand™); and the last one served as a sampling port. In each bottle,  $500 \pm 10$  mL of seawater (CaribSea Inc) were added and warmed to 40 °C. Once the temperature was reached, the  $\text{CO}_2$  cylinder (BOC 270008-J, 50 L) was opened to start saturating the seawater with respect to  $\text{CO}_2$ . The pressure was regulated by a BOC Laboratory Series 8500 Multi-stage regulator (with maximum outlet pressure of 2 bar) at 0.25 bar. The  $\text{CO}_2$  flux into each reaction bottle was regulated by a gas flowmeter (Cole-Parmer® Brass fittings, 1.2 LPM Air), and singularly controlled with gas valves (Swagelok®) positioned on the four gas lines. The  $\text{CO}_2$  injection rate was set at 0.2-0.4 L/min. To homogeneously disperse the  $\text{CO}_2$  (g) within the seawater, a stainless steel bubbler (0.5  $\mu\text{m}$  pores) was positioned at the end of each gas line. A progressive oxidation process was observed on the bubbler, raising concerns about the measurements of Fe in solution (i.e. additional dissolved Fe in solution), and required a careful cleaning after each run.

The saturation process was monitored through pH measurements, using an electrode (Fisherbrand™) connected to a pH meter (accumet™ AB15+, Fisher Scientific™). The electrode was calibrated before each sample was taken using commercial buffer solutions, and stored in a KCl solution (Fisher Chemical). Once seawater reached saturation with respect to dissolved  $\text{CO}_2$  (1-3 days from the beginning of the experiments), 80 to 30 g of ground rock was added to each of the 3 bottles. The fourth bottle was left with  $\text{CO}_2$ -saturated seawater but without any rock, in order to have a

reference chemical composition throughout the duration of the experiments. After each run, bottles, tubing, and other materials used in the experiments were cleaned with 10%  $\text{HNO}_3$  and then rinsed with DI water.

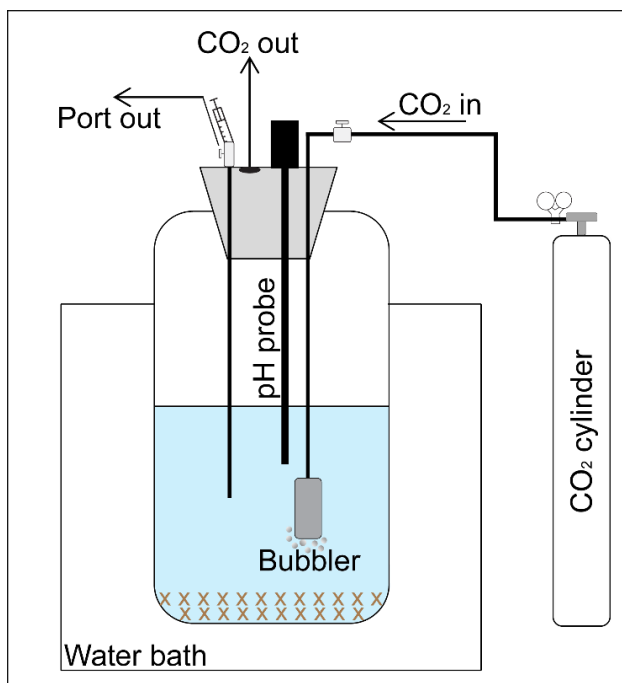


Figure 4.4: Experimental setup for the  $\text{CO}_2$ -seawater-oceanic rock dissolution experiments. The brown crosses at the bottom of the HDPE bottle represent the ground sample.

Table 4.7: Summary of the initial conditions for the  $\text{CO}_2$ -seawater-oceanic rock dissolution experiments. The reference solutions without any sample are the "Blank", and are shown at the top of each run.

| Sample         | Total Duration<br>[days] | Initial conditions |      |            |             |                                             | Saturation<br>[days] | Saturated conditions |      |            |             |                           |
|----------------|--------------------------|--------------------|------|------------|-------------|---------------------------------------------|----------------------|----------------------|------|------------|-------------|---------------------------|
|                |                          | T<br>[°C]          | pH   | TA<br>[mM] | DIC<br>[mM] | pCO <sub>2</sub><br>[ $\times 10^{-3}$ bar] |                      | T<br>[°C]            | pH   | TA<br>[mM] | DIC<br>[mM] | pCO <sub>2</sub><br>[bar] |
| Blank 2        | 11                       | 13.5               | 7.35 | 2.51       | 2.47        | 1.93                                        | 1                    | 40                   | 4.62 | 2.83       | 38.75       | 1.76                      |
| JdF mix 1      | 11                       | 13.5               | 7.35 | 2.51       | 2.47        | 1.93                                        | 1                    | 40                   | 4.63 | 2.66       | 35.54       | 1.62                      |
| JdF mix 2      | 11                       | 13.5               | 7.35 | 2.51       | 2.47        | 1.93                                        | 1                    | 40                   | 4.62 | 2.58       | 35.29       | 1.60                      |
| JdF mix 3      | 11                       | 13.5               | 7.35 | 2.51       | 2.47        | 1.93                                        | 1                    | 40                   | 4.67 | 2.95       | 36.32       | 1.63                      |
| Blank 3        | 24                       | 16.2               | 7.69 | 2.46       | 2.17        | 7.47                                        | 3                    | 40                   | 4.79 | 2.91       | 27.84       | 1.22                      |
| CD80WP132-A    | 23                       | 16.2               | 7.69 | 2.32       | 2.17        | 7.47                                        | 2                    | 40                   | 4.81 | 2.83       | 26.01       | 1.13                      |
| G1             | 24                       | 16.2               | 7.69 | 2.32       | 2.17        | 7.47                                        | 3                    | 40                   | 4.86 | 3.10       | 25.67       | 1.10                      |
| Blank 4        | 13                       | 15                 | 7.64 | 2.46       | 2.32        | 9.05                                        | 3                    | 40                   | 4.74 | 3.05       | 32.39       | 1.43                      |
| JdF mix 2bis   | 13                       | 15                 | 7.64 | 2.46       | 2.32        | 9.05                                        | 3                    | 40                   | 4.78 | 2.98       | 29.12       | 1.28                      |
| CD80WP132-Abis | 13                       | 15                 | 7.64 | 2.46       | 2.32        | 9.05                                        | 3                    | 40                   | 4.76 | 2.80       | 29.13       | 1.29                      |
| CD80WP132-B    | 13                       | 15                 | 7.64 | 2.46       | 2.32        | 9.05                                        | 3                    | 40                   | 4.77 | 2.97       | 29.63       | 1.30                      |

Fluid samples were regularly taken out directly from each bottle, opening the sampling ports one at the time, temporarily closing the CO<sub>2</sub> gas line, and using a fresh syringe for each sample. For each sampling, 1 mL of solution was taken out: 0.5 mL was immediately analysed for pH and alkalinity (Appendix “Methods” CH4-1), whereas the other 0.5 mL was kept in the refrigerator for ICP-OES and IC analyses (Appendix “Methods” CH4-2, and -3, respectively). To preserve the major cation content (Ca, K, Mg, Na, S, Si, and Sr) in the ICP-OES samples, a drop of SB conc. HNO<sub>3</sub> was added. DIC was calculated from each couple of alkalinity and pH measurements, as described in Equation 5. The corresponding pCO<sub>2</sub> was calculated using Henry’s Law for CO<sub>2</sub> solubility in water (Equation 7):

$$[CO_2] = K_0 * pCO_2 \quad (7)$$

where  $[CO_2]$  is the aqueous CO<sub>2</sub> concentration in mol/L, and  $K_0$  is Henry’s constant. The aqueous CO<sub>2</sub> concentration was calculated from carbon alkalinity (Ac) as follows:

$$[CO_2] = \frac{A_c}{(K_1/[H^+] + 2 * K_1 * K_2/[H^+])} \quad (8)$$

To determine  $K_0$  at room temperature and 40 °C, the following expression proposed by Weiss [1974] was used:

$$\ln K_0 = 9345.17/T - 60.2409 + 23.3585 \ln(T/100) + S[0.023517 - 0.0023656T + 0.0047036(T/100)^2] \quad (9)$$

where  $T$  is the temperature in Kelvin, and  $S$  is the salinity (35 psu).

Considering the initial seawater volume of  $500 \pm 15$  mL, and an average total sampling volume of 12 mL, the original rock/water ratio was not significantly modified.



### 4.3. Results

#### 4.3.1. Fluid chemistry

The variations in solution chemistry (pH, alkalinity, and DIC) as a function of time are illustrated in Figure 4.5. All the experiments show similar composition trends. The compositions of reference seawater solutions (blanks 2, 3, and 4) demonstrate how saturation conditions were reached at pH ~4.8, and alkalinity ~3.3 mmol/L, with a slight oversaturation during the first day of experiment highlighted by the calculated DIC. In general, dissolved CO<sub>2</sub> concentrations were <40 mmol/L (Figure 4.5). Experiments utilising rocks show a rapid increase in pH and alkalinity during the first five days, from initial values of 4.6 and ~2.5 mmol/L, respectively, to the maximum of 5.2 and ~5.0 mmol/L, respectively (Figure 4.5).

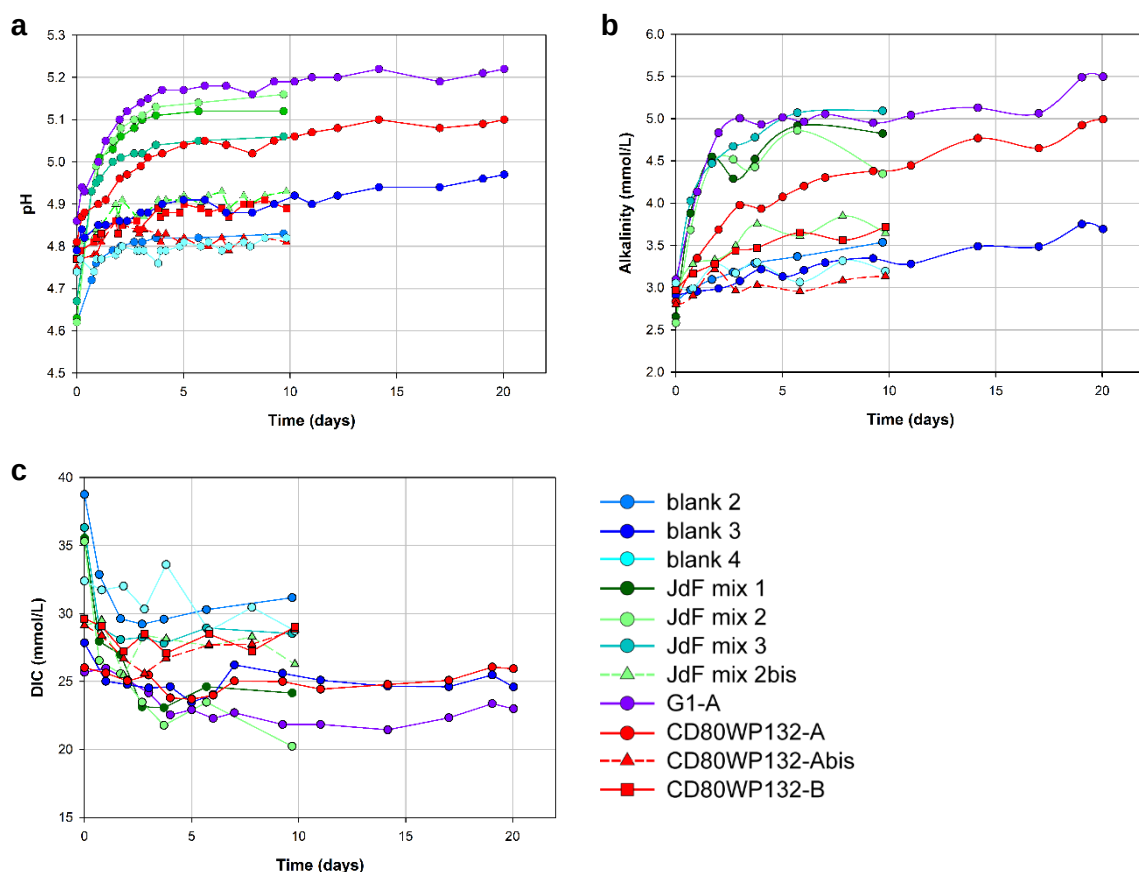


Figure 4.5: pH (a), alkalinity (b) and DIC (c) trends throughout all the duration of the experiments. For each run, the first output refers to the measurement taken at time “zero”, just before adding the solid sample, when the seawater is CO<sub>2</sub>-saturated. All the reference solutions are shown in blue and identified as blank 2, 3, and 4, respectively.

Because even the reference solutions showed an increases in pH and alkalinity, the mass change was further investigated, weighing the bottles before and after each experiment. The data, summarised in Table 4.8, confirmed the progressive evaporation of seawater during the experiments, determining a mass loss from 1 to 20% in each bottle. Assuming a linear evaporation effect, the concentrations of Na, Mg, Ca, K, Si and S in the fluids have been recalculated averaging the evaporation corrections from chlorine (Cl<sup>-</sup>), sulphate (SO<sub>4</sub><sup>2-</sup>) and total sulphur (S) analyses. The concentrations of Na, Mg, Ca, K, and S have been corrected from the “zero” day, whereas Si concentrations only from the “silicate rock addition” day because of silicon absence in seawater.

Table 4.8: Summary of evaporation effect measured via weight, anion concentrations (Cl<sup>-</sup> and SO<sub>4</sub><sup>2-</sup>), and total sulphur concentration (S), and linearized for cations and Si concentration corrections.

| Sample         | EVAPORATION (%) |                 |                               |           | MAXIMUM LINEAR EVAPORATION CORRECTION (%) |                |
|----------------|-----------------|-----------------|-------------------------------|-----------|-------------------------------------------|----------------|
|                | weight          | Cl <sup>-</sup> | SO <sub>4</sub> <sup>2-</sup> | S (total) | Average for cations                       | Average for Si |
| Blank 2        |                 | 9.52            | 8.08                          | 5.52      | 9.65                                      | 8.22           |
| JdF mix 1      |                 | 2.04            | 2.15                          | 1.92      | 1.21                                      | 1.03           |
| JdF mix 2      |                 | 4.01            | 3.78                          | 2.85      | 2.91                                      | 2.48           |
| JdF mix 3      |                 | 4.25            | 3.94                          | 1.72      | 2.58                                      | 2.19           |
| Blank 3        | 15.62           | 18.77           | 20.05                         | 13.95     | 19.51                                     | 16.00          |
| CD80WP132-A    | 9.71            | 10.09           | 9.21                          | 6.94      | 10.15                                     | 8.32           |
| G1             | 4.31            | 4.44            | 4.06                          | 3.96      | 3.92                                      | 3.21           |
| Blank 4        | 3.20            | 2.53            | 4.76                          | 3.26      | 2.64                                      | 1.93           |
| JdF mix 2bis   | 3.39            | 3.58            | 6.15                          | 3.62      | 3.56                                      | 2.61           |
| CD80WP132-Abis | 1.48            | 1.22            | 2.37                          | 1.89      | 1.56                                      | 1.14           |
| CD80WP132-B    | 5.14            | 4.70            | 6.78                          | 4.13      | 3.84                                      | 2.81           |

The original solution compositions measured with ICP-OES and IC, and the corresponding corrected concentrations, are shown in Appendix “Analyses” CH4-12.

The rock reactivity or cation release rates determined in the current study are based on the release of silicon and calcium. This choice was made for several reasons. First, both element concentrations present consistently increasing trends over time (Figure 4.6). Secondly, Si is recognised as a key element in silicate mineral dissolution reactions, because it holds together the mineral framework [Oelkers, 2001b]. Furthermore, the absence of Si and relatively low concentration of Ca in seawater make the chemical changes easier to detect in laboratory experiments. Other dissolved ions are either present in very small quantities (e.g. Al), or do not show any consistent variations in the solutions with time (e.g. Na, Mg, and K). The Ca concentrations appear to reach steady-state after ~5 days, whereas Si concentrations continue to increase throughout the experiments (Figure 4.6).

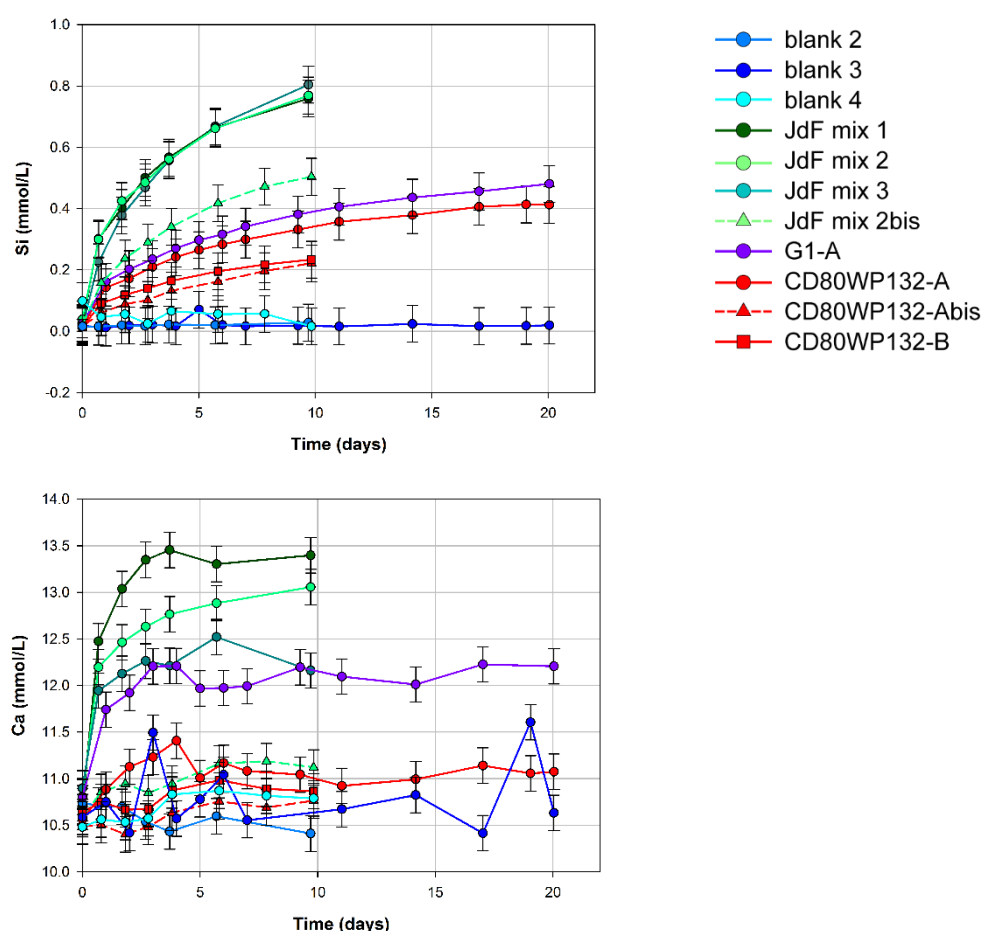


Figure 4.6: Si (top) and Ca (bottom) concentrations overtime during the dissolution experiments. Error bars represent the average 1σ (0.060 and 0.19 mmol/L for Si and Ca, respectively). For each run, the first output refers to the measurement taken at time “zero”, just before adding the rock, when the seawater is CO<sub>2</sub>-saturated. All the reference solutions are shown in blue and identified as “blank”.

Aqueous absolute (Ca/Si) and relative ( $\Delta\text{Ca}/\Delta\text{Si}$ ) ratios have been calculated for all the solutions, to better characterise the reaction stoichiometry (Figure 4.7). The Ca/Si ratio expresses the elemental concentration variations with time, whereas the  $\Delta\text{Ca}/\Delta\text{Si}$  ratio takes into account the initial solution composition, using:

$$\Delta\text{Ca}/\Delta\text{Si} = (\text{Ca}_x - \text{Ca}_0)/(\text{Si}_x - \text{Si}_0) \quad (10)$$

where “x” is the relative time in days, and “0” represents the initial concentration measured at time zero, just before adding the rock.

The Ca/Si stoichiometric ratio for basalts and gabbro used in these experiments is  $\sim 0.25$ , whereas for seawater is  $\sim 500$ .

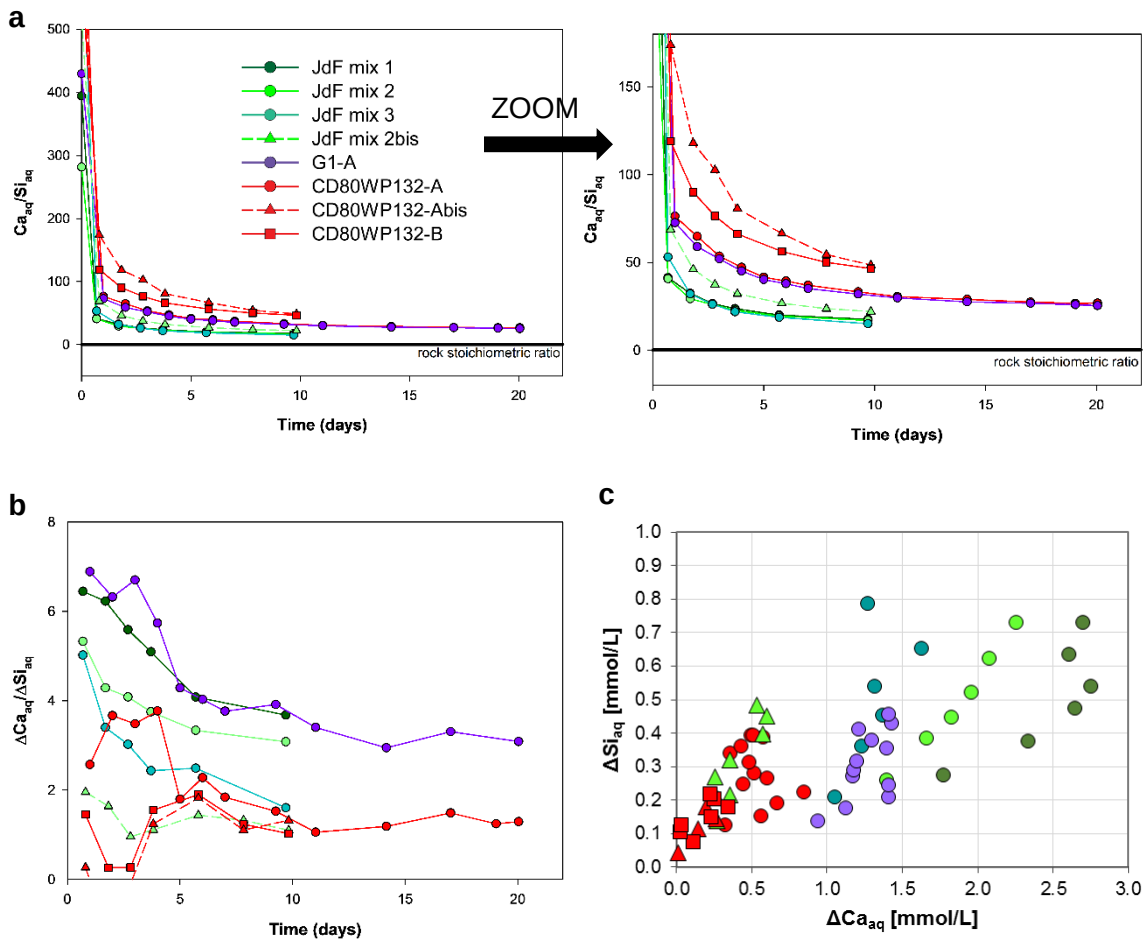


Figure 4.7: Ca/Si (a) and  $\Delta\text{Ca}/\Delta\text{Si}$  (b) ratios in all the aqueous solutions, expressed as a function of time during the dissolution experiments. Also,  $\Delta\text{Si}$  is shown as a function of  $\Delta\text{Ca}$  (c). The rock stoichiometric ratio is indicated with a black line.

Si and Ca release rates have been calculated fitting the experimental aqueous concentrations with an equation that best describes the compositional trend for each experiment (Figure 4.8), and differentiating the outputs with respect of time. These rates determine the elemental release differences between the samples, and quantify the incremental Si and Ca concentrations in solution over time (Figure 4.9). The R-squared coefficient, which range from 0 (the model fits no data) to 1 (the model fits all data), is used to described how well the data are fitted by the regression curve (Table 4.9). For the silicon concentrations all R-squared coefficients are >0.97, showing a very good agreement between curve and data. For the calcium concentrations, which show more variability among the samples, R-squared is >0.89 for JdF samples, 0.45< R-squared >0.77 for CD80WP132 samples, and >0.91 for G1.

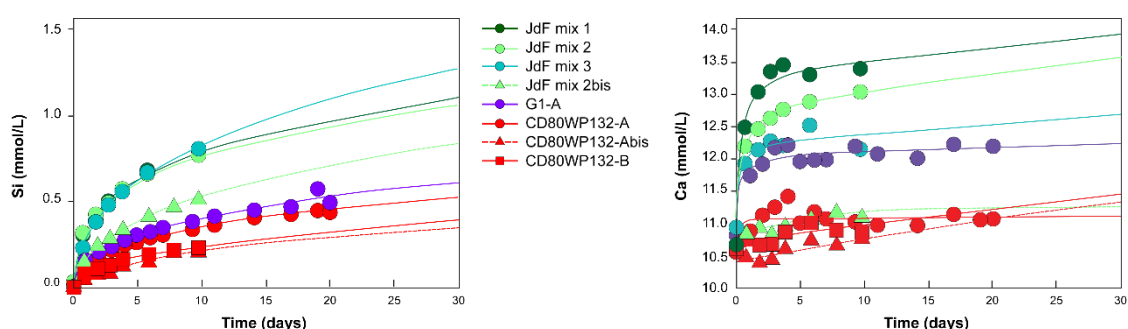


Figure 4.8: Best fitting curves of silicon (left) and calcium (right) concentrations for all the experimental runs.

Table 4.9: Summary of the R-squared coefficients for the regression curves applied to silicon and calcium concentrations in aqueous solution, for all the experimental run.

|                | R-squared |         |
|----------------|-----------|---------|
|                | Silicon   | Calcium |
| JdF mix 1      | 0.9976    | 0.9813  |
| JdF mix 2      | 0.9990    | 0.9986  |
| JdF mix 2bis   | 0.9986    | 0.8941  |
| JdF mix 3      | 0.9984    | 0.9464  |
| G1-A           | 0.9798    | 0.9156  |
| CD80WP132-A    | 0.9966    | 0.4589  |
| CD80WP132-Abis | 0.9974    | 0.7629  |
| CD80WP132-B    | 0.9979    | 0.6567  |

The differentiation has been done considering that the best fit is given by an exponential function at power two in every case (Equation 11, and its derivative Equation 12):

$$f(x) = ax^b + c \quad (11)$$

$$f'(x) = ab x^{b-1} \quad (12)$$

where  $f(x)$  is the element concentration in mmol/L,  $x$  is the time ranging from 0 up to 30 days,  $a, b$ , and  $c$  the constant values obtained with the best fit equation for each experiment, and  $f'(x)$  the element release rate in mmol/L/day. The results of both Ca and Si release rates are shown in Figure 4.9. During the first two days the release rates range from 0.24 to 0.04 mmol/L/day for Si, and from 1.07 to 0.03 mmol/L/day for Ca. After 5 days, both Si and Ca release rates asymptotically decrease towards values  $<0.02$  mmol/L/day. In general, the Si release rates among all the samples show similar trends, whereas the calcium ones display more variability.

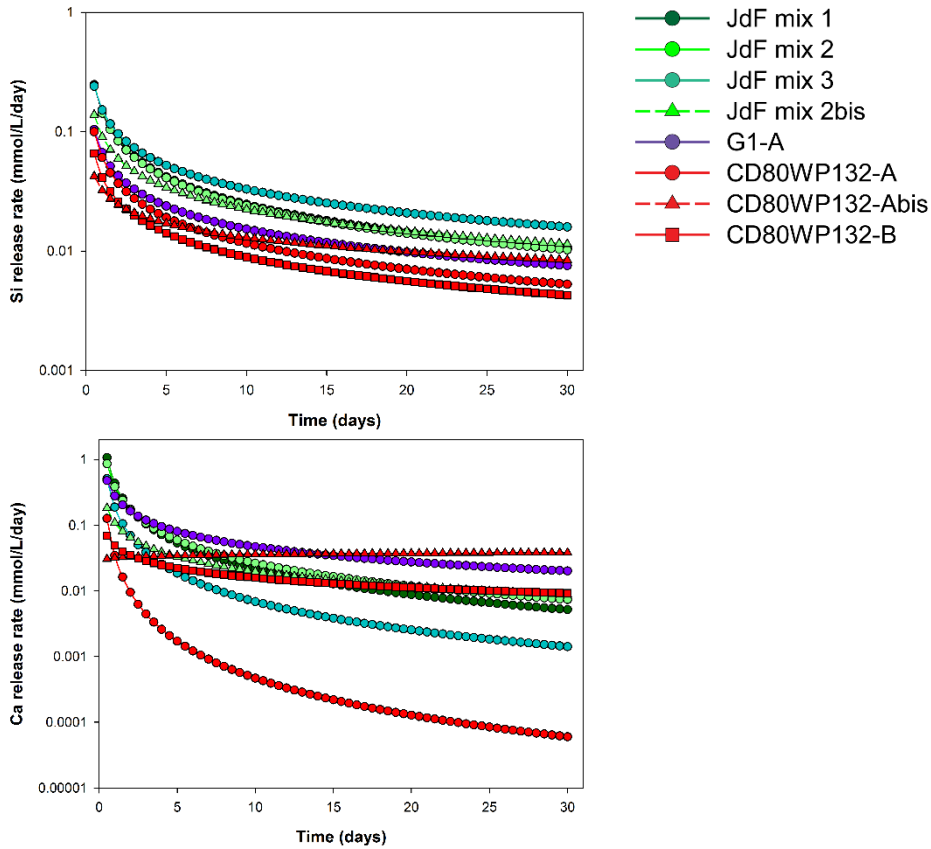


Figure 4.9: Si (top) and Ca (bottom) release rate as a function of time during the dissolution experiments.

### 4.3.2. Fine particle removal

Figures 4.10 (for JdF) and 4.11 (for G1 and CD80WP132) show GSD measurements, and the associated SEM images, for the samples before and after the experiments.

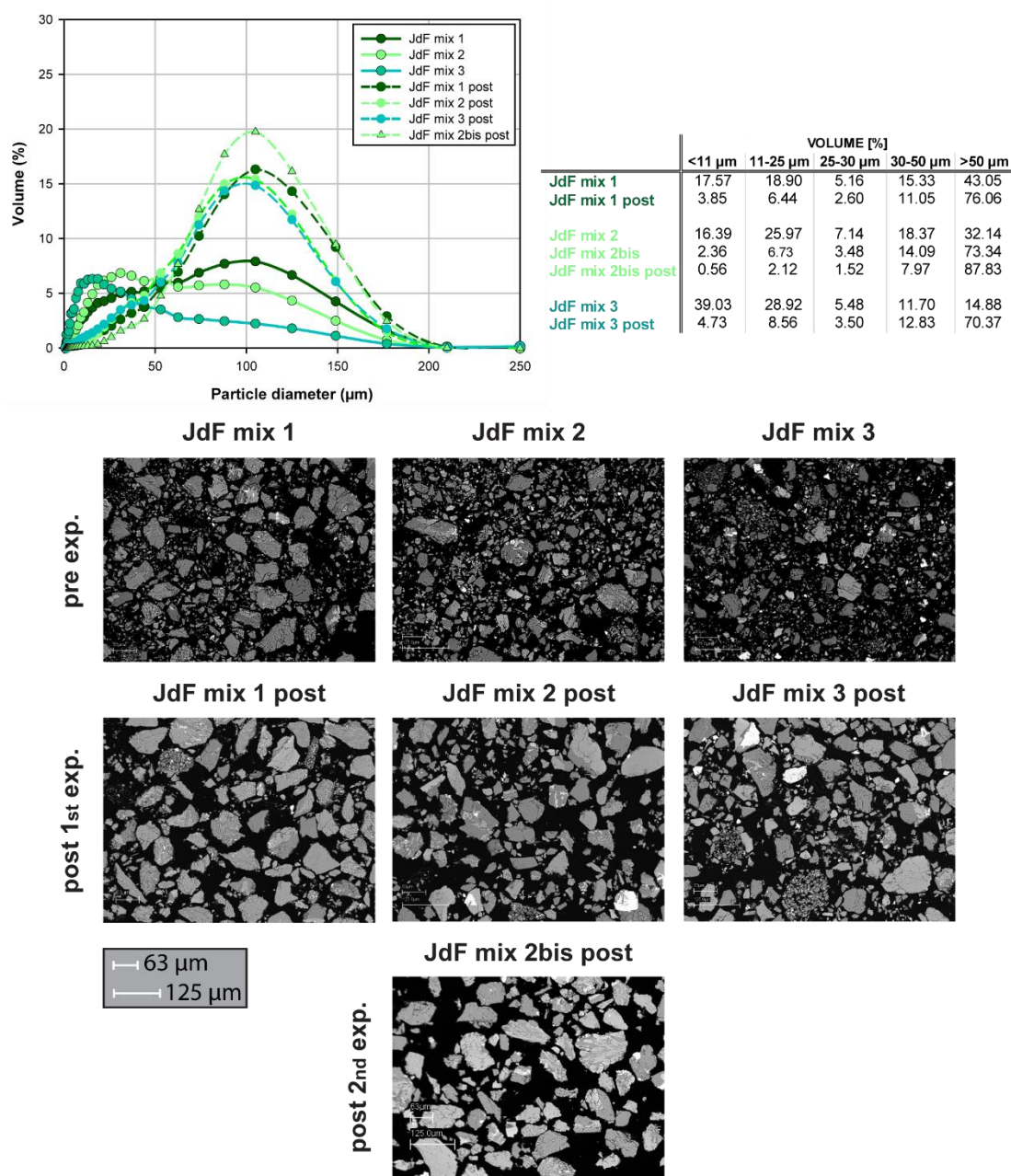


Figure 4.10: Grain size distribution analyses (top) and SEM images (bottom) for all the JdF samples, before and after the experiments (pre and post, respectively). The SEM images were taken with 500x of magnification. During the first set of experiments, 80 g of material was used for each run, whereas during the second set 65 g of already reacted material.



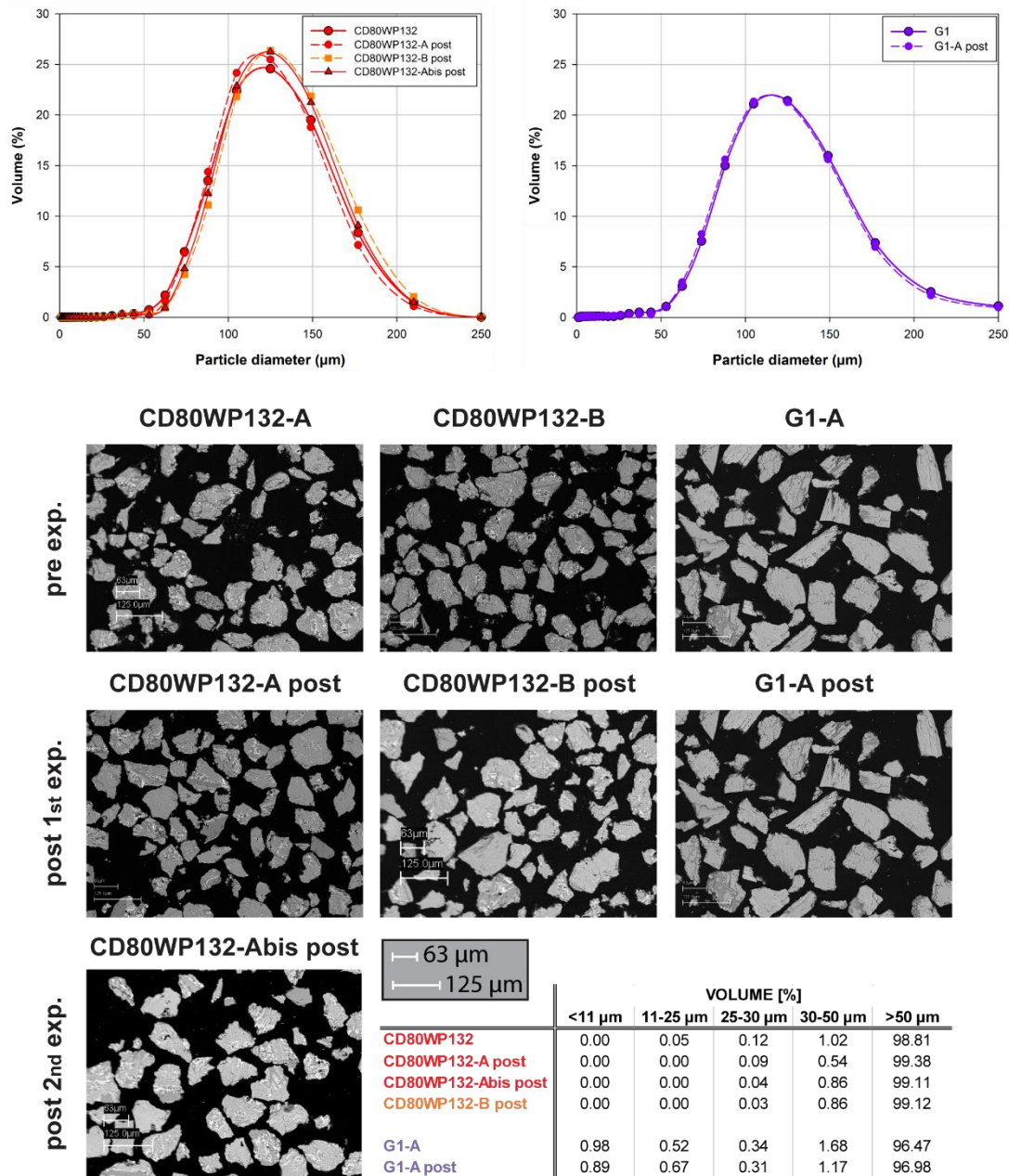


Figure 4.11: Grain size distribution analyses (top) and SEM images (bottom) for all the CD80WP132 basalts (left) and G1 gabbro (right), before and after the experiments (pre and post, respectively). The SEM images were taken with 500x of magnification. During the first set of experiments, 80 g of material was used for each run, whereas during the second set 65 g of already reacted material was used. The experimental run on the sample CD80WP132-B was executed with only 30 g.

Importantly, the absence of fine particles ( $<10 \mu\text{m}$ ) in G1 and CD80WP132 confirms that the dissolution of these samples has to be related to the effective size fraction between  $63\text{--}125 \mu\text{m}$ , for all the experiments (Figure 4.11). Different scenarios are presented by the analyses of JdF samples. Both GSD and SEM analyses on samples JdF mix 1, 2,



and 3 before the dissolution experiments indicate the presence of 16 to 40% in volume of fine particles. Analyses of the corresponding samples that underwent the first experimental run show a decrease in clay-sized material of 2-5% in volume (Figure 4.10, JdF mix 1, 2, and 3 post). This means that the fast kinetics expressed by Si and Ca release rates of JdF samples are most probably related to their higher reactive surface area, in agreement with the observations in previous studies [Andreani *et al.*, 2009; Gadikota *et al.*, 2014b; Helgeson *et al.*, 1984]. On the other hand, the second experimental run on the already used JdF mix 2 sample does not have such an abundance of fine particles, and the dissolution rates measured can be directly compared with other experiments with similar grain size fraction between 63-125  $\mu\text{m}$  (Figure 4.10, JdF mix 2bis post).

Considering the change in grain size as a critical parameter in the kinetic rate interpretation, particle diameters and corresponding geometric surface areas ( $A_{\text{geo}}$ ) of each sample are summarised in Table 4.10, before and after the dissolution experiments.

Table 4.10: Particle diameters and correspondent geometric surface areas ( $A_{\text{geo}}$ ) for each sample, before and after the dissolution experiments. The elapsed time is specified.

| Sample                | PRE EXPERIMENT            |                                            | Elapsed time<br>days | POST EXPERIMENT           |                                            |
|-----------------------|---------------------------|--------------------------------------------|----------------------|---------------------------|--------------------------------------------|
|                       | GSD mean<br>$\mu\text{m}$ | $A_{\text{geo}}$<br>$\text{cm}^2/\text{g}$ |                      | GSD mean<br>$\mu\text{m}$ | $A_{\text{geo}}$<br>$\text{cm}^2/\text{g}$ |
| <b>JdF mix 1</b>      | 50.95                     | 406                                        | 9.7                  | 79.44                     | 260                                        |
| <b>JdF mix 2</b>      | 42.35                     | 489                                        | 9.7                  | 74.05                     | 279                                        |
| <b>JdF mix 2bis</b>   | 74.05                     | 279                                        | 9.8                  | 86.88                     | 238                                        |
| <b>JdF mix 3</b>      | 26.50                     | 781                                        | 9.7                  | 72.61                     | 285                                        |
| <b>G1</b>             | 108.44                    | 191                                        | 20.0                 | 107.25                    | 193                                        |
| <b>CD80WP132-A</b>    | 109.89                    | 188                                        | 20.0                 | 109.82                    | 188                                        |
| <b>CD80WP132-Abis</b> | 109.82                    | 188                                        | 9.8                  | 112.84                    | 183                                        |
| <b>CD80WP132-B</b>    | 103.72                    | 199                                        | 9.8                  | 114.94                    | 180                                        |

### 4.3.3. Rock dissolution

The term “rock dissolution” refers to the maximum degree of dissolution to which a rock is characterised by. Usually, this level is reached with the steady-state ion release rate, which is defined as a condition where dissolution rates are time independent and where dissolution is stoichiometric [Oelkers, 2001b]. Although an almost constant pH is reached after five days in all the experimental runs, the steady-state condition is not fully reached in this study (Section 4.4.2).

The extent of rock dissolution is shown as a function of time in Figure 4.12 and summarised in Table 4.11. The percentages have been calculated as a ratio between aqueous concentration difference, and whole rock content, for both Si and Ca. The data for the whole rock content have been taken from XRF and SEM-EDS bulk rock analyses. All the mass of the samples used during the experiment is considered as “reactive”.

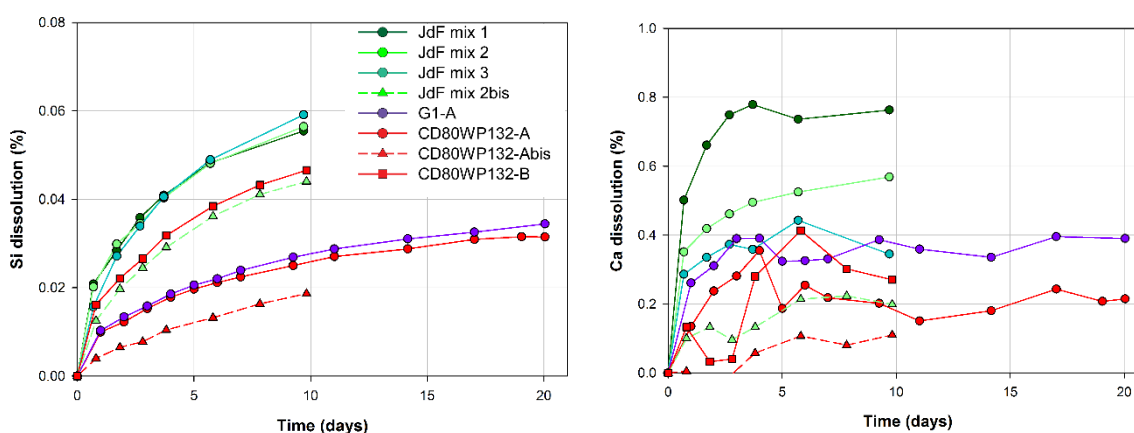


Figure 4.12: Rock dissolution extents (Si and Ca) in % vs. time for all the samples used in the experiments. All the mass is considered as “reactive”.

In general, the greatest extent of dissolution in terms of mass lost was achieved in the JdF samples, rich in fine particles, followed by the ophiolitic gabbro. The lowest extent of dissolution was observed in the MAR sample subjected to consecutive experimental runs (CD80WP132-Abis). The sample JdF mix 2bis (lower fine particle content) shows intermediate dissolution extent similar to fresh MAR basalt (CD80WP132-A and B). In

particular, the comparable calcium dissolution extents of CD80WP132-A and B, which weigh 80 and 30 g respectively, highlight the apparent non-dependence on sample mass. This observation leads to the hypothesis that the reactivity is focused only on the top-surface layer of rock samples in the reactors (further explanations in Section 4.4.2.2).

*Table 4.11: Rock dissolution extent in % using Si and Ca concentrations for all the samples used during the experiment. All the rock is considered to be “reactive”.*

| Sample         | Mass<br>[g] | Type   | Elapsed<br>time<br>[days] | AQUEOUS CONCENTRATIONS                   |                                          | ROCK DISSOLUTION |           |
|----------------|-------------|--------|---------------------------|------------------------------------------|------------------------------------------|------------------|-----------|
|                |             |        |                           | $\Delta\text{Si}$<br>in 500 mL<br>[mmol] | $\Delta\text{Ca}$<br>in 500 mL<br>[mmol] | Si<br>[%]        | Ca<br>[%] |
| JdF mix 1      | 80          | fresh  | 9.7                       | 0.37                                     | 1.35                                     | 0.06             | 0.76      |
| JdF mix 2      | 80          | fresh  | 9.7                       | 0.37                                     | 1.13                                     | 0.06             | 0.57      |
| JdF mix 2bis   | 65          | re-run | 9.8                       | 0.24                                     | 0.27                                     | 0.04             | 0.20      |
| JdF mix 3      | 80          | fresh  | 9.7                       | 0.39                                     | 0.63                                     | 0.06             | 0.35      |
| G1-A           | 80          | fresh  | 20.4                      | 0.23                                     | 0.70                                     | 0.03             | 0.39      |
| CD80WP132-A    | 80          | fresh  | 20.4                      | 0.20                                     | 0.26                                     | 0.03             | 0.22      |
| CD80WP132-Abis | 65          | re-run | 9.8                       | 0.10                                     | 0.14                                     | 0.02             | 0.11      |
| CD80WP132-B    | 30          | fresh  | 9.8                       | 0.11                                     | 0.11                                     | 0.05             | 0.27      |
| Average        | -           | -      | -                         | -                                        | -                                        | 0.04             | 0.33      |

## 4.4. Discussion

### 4.4.1. Characterisation of physical properties

The surface area available for reaction is one of the most important factors determining the rate and extent of reaction for a given rock volume and time. Fine particles (<10  $\mu\text{m}$ ) have a higher ratio of surface area to volume, which in turn leads to high rock dissolution rates. Therefore, it is essential to carefully characterise the particle size and the corresponding reactive surface area in order to avoid overestimating the dissolution rates.

To address this issue, a clear procedure to prepare the samples was defined [Gadikota *et al.*, 2014b], and analyses on the grain size distribution (GSD) and on the surface area (GSA and BET) have been carried out. In general, BET analyses with nitrogen give a

good measurement of specific surface area higher than  $1 \text{ m}^2/\text{g}$ , whereas krypton is more suitable for specific surface area  $<1 \text{ m}^2/\text{g}$  [Lowell and Shields, 2013]. However, BET surface area may be a poor estimate for water-rock experiments because it is not representative of the actual contact area between fluid and rocks [Brantley and Mellott, 2000; Lüttge and Arvidson, 2008]. On the other hand, surface roughness is not taken into account in the geometric surface area calculations, which assume the presence of smooth spheres [Cubillas *et al.*, 2005]. Many studies have demonstrated how the geometric rather than BET surface area may be more accurate for estimating dissolution rates [Gautier *et al.*, 2001; Gysi and Stefánsson, 2012c; Wolff-Boenisch and Gíslason, 2002; Wolff-Boenisch *et al.*, 2004; Wolff-Boenisch *et al.*, 2011]. Rates in this study were normalized to the geometric surface area (Section 4.4.2), which was found to be more consistent with the SEM observations on particle size than BET measurements carried out with nitrogen (Table 4.6).

#### **4.4.2. Rock dissolution characterisation**

The interaction of  $\text{CO}_2$ -rich seawater and mafic rocks results in changes in solution chemistry, that are dependent on several factors including the  $\text{CO}_2$  concentration, temperature, rock mineralogy, and sample grain size.

With respect to  $\text{CO}_2$  sequestration, the primary interests of this experimental study are related to the determination of the rock dissolution rates (Section 4.4.2.1), the reactive mass within the sample – top-surface only hypothesis (Section 4.4.2.2), the type of mass transfer from rock to solution – congruent and/or incongruent dissolution (Section 4.4.2.3), and the calcium sources within the rocks (Section 4.4.2.4).

#### 4.4.2.1. Rock dissolution rates

Considering the grain size distribution differences between the samples, the dissolution rates have been normalised to the geometric surface area (Figure 4.13), following the equation:

$$r_X = \log [\Delta X / (A_{\text{geo}} * t)] \quad (13)$$

where  $r_X$  is the dissolution rate of element  $X$  expressed as  $\log [\text{mol}/\text{cm}^2/\text{s}]$ ,  $\Delta X$  is the differential aqueous concentration of element  $X$  in  $\text{mol}/\text{kg}$ ,  $A_{\text{geo}}$  the sample-specific geometric surface area pre-experiment in  $\text{cm}^2/\text{kg}$  (Table 4.10), and  $t$  the elapsed time in seconds [Gysi and Stefánsson, 2012c].

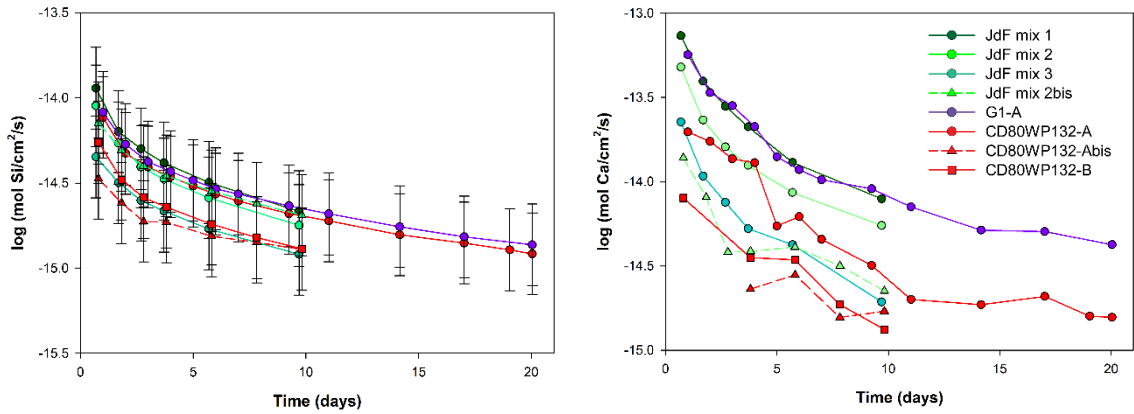


Figure 4.13: Si (on the left) and Ca (on the right) dissolution rates as a function of time normalized to the initial geometric surface area of rock for all experimental runs. Ca dissolution rates for CD80WP132-Abis (days 0.8, 1.8, and 2.8) and CD80WP132-B (days 1.8 and 2.8) have not been plotted due to analytical errors (values available in Table 4.12). Error bars (Si) or size of symbols (Ca) correspond to a  $\pm 0.2$  or  $\pm 0.01$  log unit estimated uncertainty calculated from the precision of Si and Ca concentrations measurements, respectively (Appendix “Methods” CH4-2, IAPSO error).

The dissolution rates in this study decrease with time, and trend towards the rock stoichiometric dissolution rate (Figure 4.7). In general, Si dissolution rates range from -13.94 to -14.92  $\log (\text{mol}/\text{cm}^2/\text{s})$ , whereas Ca dissolution rates range from -13.13 to -14.88  $\log (\text{mol}/\text{cm}^2/\text{s})$  (Table 4.12). The dissolution rates for Si and Ca measured at the end of each experimental run are the closest to the stoichiometric dissolution, and range from -14.67 to -14.92  $\log (\text{mol}/\text{cm}^2/\text{s})$ , and from -14.10 to -14.88  $\log (\text{mol}/\text{cm}^2/\text{s})$ , respectively (Table 4.12).

The similarity among the dissolution rates confirms that the fastest kinetics related to fine particle abundance are overcome by the normalisation, making possible an appropriate comparison among samples. Overall, the ophiolitic gabbro shows dissolution rates higher or similar to the JdF ones. The MAR basalts are characterised by a wider range of dissolution rates that are usually lower than JdF and gabbroic samples.

Table 4.12: Summary of rock dissolution rates, normalised to the geometric surface area, for all the experimental runs. Values of dissolution rate are given in log (mol/cm<sup>2</sup>/s). For CD80WP132-Abis, the Ca dissolution rates for days 1.8 and 2.8 are not available due to analytical errors on Ca concentrations.

| Elapsed time (days) | JdF mix 1 |        | JdF mix 2 |        | JdF mix 2bis |        | JdF mix 3 |        | G1     |        | CD80WP132 -A |        | CD80WP132 -Abis |        | CD80WP132 -B |        |
|---------------------|-----------|--------|-----------|--------|--------------|--------|-----------|--------|--------|--------|--------------|--------|-----------------|--------|--------------|--------|
|                     | Si        | Ca     | Si        | Ca     | Si           | Ca     | Si        | Ca     | Si     | Ca     | Si           | Ca     | Si              | Ca     | Si           | Ca     |
| 0.7-0.8             | -13.94    | -13.13 | -14.05    | -13.32 | -14.15       | -13.86 | -14.35    | -13.65 | -      | -      | -            | -      | -14.47          | -15.05 | -14.26       | -14.10 |
| 1.0                 | -         | -      | -         | -      | -            | -      | -         | -      | -14.09 | -13.25 | -14.12       | -13.71 | -               | -      | -            | -      |
| 1.7-1.8             | -14.20    | -13.40 | -14.27    | -13.63 | -14.31       | -14.09 | -14.50    | -13.97 | -      | -      | -            | -      | -14.62          | -      | -14.48       | -15.06 |
| 2.0                 | -         | -      | -         | -      | -            | -      | -         | -      | -14.27 | -13.47 | -14.33       | -13.76 | -               | -      | -            | -      |
| 2.7-2.8             | -14.30    | -13.55 | -14.41    | -13.79 | -14.40       | -14.42 | -14.60    | -14.12 | -      | -      | -            | -      | -14.73          | -      | -14.59       | -15.16 |
| 3.0                 | -         | -      | -         | -      | -            | -      | -         | -      | -14.38 | -13.55 | -14.41       | -13.86 | -               | -      | -            | -      |
| 3.7-3.8             | -14.38    | -13.68 | -14.48    | -13.90 | -14.46       | -14.41 | -14.67    | -14.28 | -      | -      | -            | -      | -14.73          | -14.64 | -14.64       | -14.45 |
| 4.0                 | -         | -      | -         | -      | -            | -      | -         | -      | -14.43 | -13.67 | -14.46       | -13.89 | -               | -      | -            | -      |
| 5.0                 | -         | -      | -         | -      | -            | -      | -         | -      | -14.48 | -13.85 | -14.52       | -14.26 | -               | -      | -            | -      |
| 5.7-5.8             | -14.50    | -13.89 | -14.59    | -14.06 | -14.55       | -14.39 | -14.77    | -14.37 | -      | -      | -            | -      | -14.81          | -14.55 | -14.74       | -14.46 |
| 6.0                 | -         | -      | -         | -      | -            | -      | -         | -      | -14.53 | -13.93 | -14.57       | -14.21 | -               | -      | -            | -      |
| 7.0                 | -         | -      | -         | -      | -            | -      | -         | -      | -14.56 | -13.99 | -14.61       | -14.34 | -               | -      | -            | -      |
| 7.8                 | -         | -      | -         | -      | -14.62       | -14.50 | -         | -      | -      | -      | -            | -      | -14.85          | -14.81 | -14.82       | -14.73 |
| 9.3                 | -         | -      | -         | -      | -            | -      | -         | -      | -14.63 | -14.04 | -14.68       | -14.50 | -               | -      | -            | -      |
| 9.7-9.8             | -14.67    | -14.10 | -14.75    | -14.26 | -14.69       | -14.65 | -14.92    | -14.71 | -      | -      | -            | -      | -14.89          | -14.77 | -14.89       | -14.88 |
| 11.0                | -         | -      | -         | -      | -            | -      | -         | -      | -14.68 | -14.15 | -14.72       | -14.70 | -               | -      | -            | -      |
| 14.2                | -         | -      | -         | -      | -            | -      | -         | -      | -14.76 | -14.29 | -14.80       | -14.73 | -               | -      | -            | -      |
| 17.0                | -         | -      | -         | -      | -            | -      | -         | -      | -14.82 | -14.30 | -14.85       | -14.68 | -               | -      | -            | -      |
| 19.0                | -         | -      | -         | -      | -            | -      | -         | -      | -      | -      | -14.89       | -14.80 | -               | -      | -            | -      |
| 20.0                | -         | -      | -         | -      | -            | -      | -         | -      | -14.86 | -14.37 | -14.92       | -14.81 | -               | -      | -            | -      |

#### 4.4.2.2. Reactive mass

For a fixed elemental concentration in solution, the key parameter to calculate the dissolution extent is the effective reactive mass in solution.

In section 4.3.3, a general estimation of dissolution extent is given, considering all the sample as reactive. However, it has been observed that a change in the sample mass used during the experiments (from 80 to 30 g) does not significantly affect the Ca and Si concentrations in solution. Therefore, the hypothesis has been advanced that the

reactions are only focused on the top-surface layer of rock samples in the reactor. If this hypothesis is true, the dissolution extent should be higher, depending on the size of the top-surface area considered as reactive (Figure 4.14).

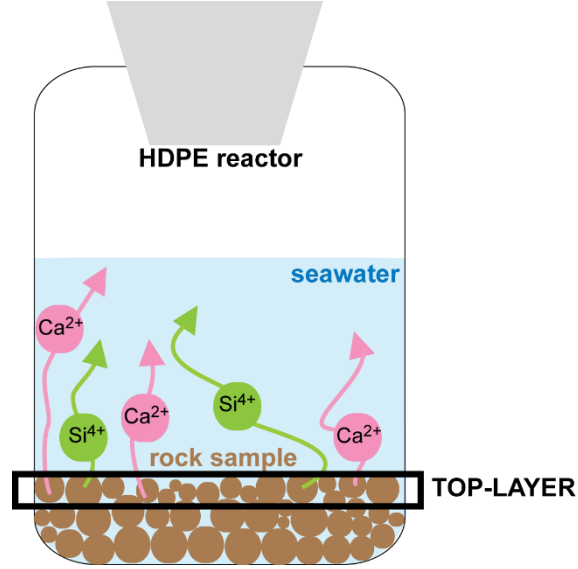


Figure 4.14: Sketch of rock dissolution within the HDPE reactor used in this study, considering only the top-surface area of the sample ("top-layer") as reactive.

An attempt has been made to quantify the top-layer weight (Equation 14), assuming the presence of unimodal grain size with diameter equal to the mean of the corresponding grain size distribution for each sample (Table 4.10).

$$M_{TL} = (A_{HDPE} * V_p * \rho_r) / A_p \quad (14)$$

where  $M_{TL}$  is the top-layer mass in g,  $A_{HDPE}$  the HDPE bottle area calculated for an actual radius of 4 cm,  $V_p$  and  $A_p$  the volume and area, respectively, occupied by a single particle (accordingly to GSD mean), and  $\rho_r$  the rock density assumed to be 2.9 g/cm<sup>3</sup>.

On average, the top-layer weighs 0.76 g. Also, assuming 50 wt% SiO<sub>2</sub> (Si = 23.4 wt%) and 12 wt% CaO (Ca = 8.6 wt%) for each layer (averages for mafic rocks), it is possible

to estimate the maximum Si and Ca release if the top-layer totally reacted, and compare it with the actual Si and Ca aqueous concentrations (Table 4.13). Calculations show that an average reactive mass of 0.76 g, lying on the top-surface always in contact with the CO<sub>2</sub>-rich solution, is sufficient to supply the measured concentrations during the dissolution experiments. This means that the average values of Si and Ca dissolution extent for the whole sample mass of 0.04% and 0.33%, respectively (Table 4.11), would increase two orders of magnitude if all reactions are taking place from the top-surface only (Table 4.13). However, these preliminary considerations have to be further tested and confirmed by more experiments with different water/rock ratios, and stirred samples, in order to better constrain the dissolution extent and the reactive mass of these rocks.

Table 4.13: Summary of top-surface reactive mass ( $M_{TL}$ ) for all the experimental runs, with corresponding Si and Ca mass ( $M_{Si-TL}$  and  $M_{Ca-TL}$ , respectively).  $\Delta Si(aq)$  and  $\Delta Ca(aq)$  are the differential moles of Si and Ca, respectively, measured in 500 mL of aqueous solution (Table 4.11).  $Si_{TL}$  and  $Ca_{TL}$  are the Si and Ca rock dissolution extent, respectively, considering only the top-surface area as reactive. Grey shading shows the sample averages for all the above mentioned parameters.

| Sample                | $M_{TL}$<br>[g] | $M_{Si-TL}$<br>[g] | $Mol_{Si-TL}$<br>[mol] | $\Delta Si(aq)$<br>[mol] | $Si_{TL}$<br>[%] | $M_{Ca-TL}$<br>[g] | $Mol_{Ca-TL}$<br>[mol] | $\Delta Ca(aq)$<br>[mol] | $Ca_{TL}$<br>[%] |
|-----------------------|-----------------|--------------------|------------------------|--------------------------|------------------|--------------------|------------------------|--------------------------|------------------|
| JdF mix 1             | 0.50            | 0.12               | 0.0041                 | 0.0004                   | 5.84             | 0.04               | 0.0011                 | 0.0014                   | 80.34            |
| JdF mix 2             | 0.41            | 0.10               | 0.0034                 | 0.0004                   | 5.94             | 0.04               | 0.0009                 | 0.0011                   | 59.87            |
| JdF mix 3             | 0.26            | 0.06               | 0.0021                 | 0.0004                   | 6.23             | 0.02               | 0.0006                 | 0.0006                   | 36.32            |
| JdF mix 2bis          | 0.72            | 0.17               | 0.0060                 | 0.0002                   | 3.77             | 0.06               | 0.0015                 | 0.0003                   | 15.88            |
| Average (JdF)         | 0.47            | 0.11               | 0.0039                 | 0.0035                   | 5.44             | 0.04               | 0.0010                 | 0.0009                   | 48.10            |
| G1-A                  | 1.05            | 0.25               | 0.0088                 | 0.0002                   | 3.63             | 0.09               | 0.0023                 | 0.0007                   | 41.06            |
| CD80WP132-A           | 1.07            | 0.25               | 0.0089                 | 0.0002                   | 3.31             | 0.09               | 0.0023                 | 0.0003                   | 22.64            |
| CD80WP132-B           | 1.01            | 0.24               | 0.0084                 | 0.0001                   | 1.84             | 0.09               | 0.0022                 | 0.0001                   | 9.96             |
| CD80WP132-Abis        | 1.07            | 0.25               | 0.0089                 | 0.0001                   | 1.59             | 0.09               | 0.0023                 | 0.0001                   | 8.79             |
| Average (CD80WP132)   | 1.05            | 0.25               | 0.0087                 | 0.0001                   | 2.25             | 0.09               | 0.0023                 | 0.0002                   | 13.79            |
| Average (all samples) | 0.76            | 0.18               | 0.0063                 | 0.0003                   | 4.02             | 0.07               | 0.0016                 | 0.0006                   | 34.36            |

#### 4.4.2.3. Rock mass transfer

As shown in Figure 4.7, the extent of dissolution in terms of Ca/Si ratio decreases with rock dissolution progress, getting closer – and in some cases parallel – to the whole rock Ca/Si stoichiometric ratio (~0.25), but without reaching it. This means that all the experimental runs are characterised by incongruent dissolution. *Oelkers* [2001b]



explained the “non-stoichiometry” as a result of the leached layer formation, which is commonly a provisional part of the dissolution process. Observations of this layer on olivine show that the leaching of cations and incongruent mineral dissolution result in the formation of a Si-rich mass-transfer-limiting passivation layer, which eventually limits the extent of mineral dissolution [Béarat *et al.*, 2006; Daval *et al.*, 2011; Gadikota *et al.*, 2014b]. In this study, the formation of a passivation layer is suggested by the observations of incongruent dissolution, and of a higher calcium dissolution extent than Si one (Tables 4.11 and 4.13). Also, the proximity of Ca and Si release rates to steady-state indicates that the reactions are extending beyond the passivation layer to the bulk rock, towards a more congruent dissolution (Figure 4.9).

#### 4.4.2.4. Calcium source

The different release of calcium at pH ~5 from all the mafic rocks used during the experiments (Tables 4.11 and 4.13) suggests which of the calcium-rich minerals are the most susceptible to dissolution. In the rocks used in this study, calcium is abundant in glass, mesostasis, pyroxene, amphibole, plagioclase, and calcite (Table 4.14).

Table 4.14: Averages of mineral molar compositions [Deer *et al.*, 1963; 1978; 1997], and relative stoichiometric Ca/Si ratios. Glass composition from USGS standard BHVO-2 [Wilson, 1997].

|                                 | Ca  | Na  | Mg  | Fe  | Ti   | Al  | Si  | C   | O   | OH | Ca/Si |
|---------------------------------|-----|-----|-----|-----|------|-----|-----|-----|-----|----|-------|
|                                 | mol |     |     |     |      |     |     |     |     |    |       |
| <b>Plagioclase, albite</b>      | -   | 1.0 | -   | -   | -    | 1.0 | 3.0 | -   | 8.0 | -  | -     |
| <b>Plagioclase, oligoclase</b>  | 0.2 | 0.8 | -   | -   | -    | 1.2 | 2.8 | -   | 8.0 | -  | 0.07  |
| <b>Plagioclase, andesine</b>    | 0.4 | 0.6 | -   | -   | -    | 1.4 | 2.6 | -   | 8.0 | -  | 0.15  |
| <b>Plagioclase, labradorite</b> | 0.6 | 0.4 | -   | -   | -    | 1.6 | 2.4 | -   | 8.0 | -  | 0.25  |
| <b>Plagioclase, bytownite</b>   | 0.8 | 0.2 | -   | -   | -    | 1.8 | 2.2 | -   | 8.0 | -  | 0.36  |
| <b>Plagioclase, anorthite</b>   | 1.0 | -   | -   | -   | -    | 2.0 | 2.0 | -   | 8.0 | -  | 0.50  |
| <b>Pyroxene, augite</b>         | 0.9 | 0.1 | 0.9 | 0.2 | 0.10 | 0.4 | 1.9 | -   | 6.0 | -  | 0.42  |
| <b>Amphibole, actinolite</b>    | 1.8 | 0.2 | 3.4 | 1.3 | 0.02 | 0.4 | 7.8 | -   | 22  | 2  | 0.23  |
| <b>Calcite</b>                  | 1.0 | -   | -   | -   | -    | -   | -   | 1.0 | 3.0 | -  | -     |
| <b>Glass</b>                    | 0.2 | 0.1 | 0.2 | 0.2 | 0.03 | 0.3 | 0.8 | -   | 2.9 | -  | 0.25  |

In order to advance suggestions on which is the most reactive mineral among them, the mineral dissolution rates have to be compared. According to literature data, calcite, glass and olivine are the most susceptible phases to alteration, followed by pyroxene, amphibole, and plagioclase [e.g., *Banfield et al.*, 1991; *Crovisier et al.*, 1987; *Eggleton et al.*, 1987]. However, the alteration order of the last three minerals is not well established, because it mainly depends on pH, temperature, and mineral composition [e.g., *Fridriksson et al.*, 2001; *Gudbrandsson et al.*, 2011; *Nesbitt and Wilson*, 1992]. A list of mineral dissolution rates, together with the experimental conditions at which they have been measured, is available in Table 4.15. In general, these kinetic observations suggest that, for pH ~5, the mineral alteration order is calcite > olivine and glass > pyroxene > anorthitic plagioclase > amphibole hornblende > albitic plagioclase.

Given the mineral average distributions of the samples used in this study (Figure 4.15), there are several Ca-rich phases to be considered as potential Ca-sources. For the basalts, these phases are calcite, mesostasis, labradoritic plagioclase, and augitic pyroxene, with the addition of glass for the MAR samples, whereas for the ophiolitic gabbro G1, amphibole and anorthitic plagioclase.

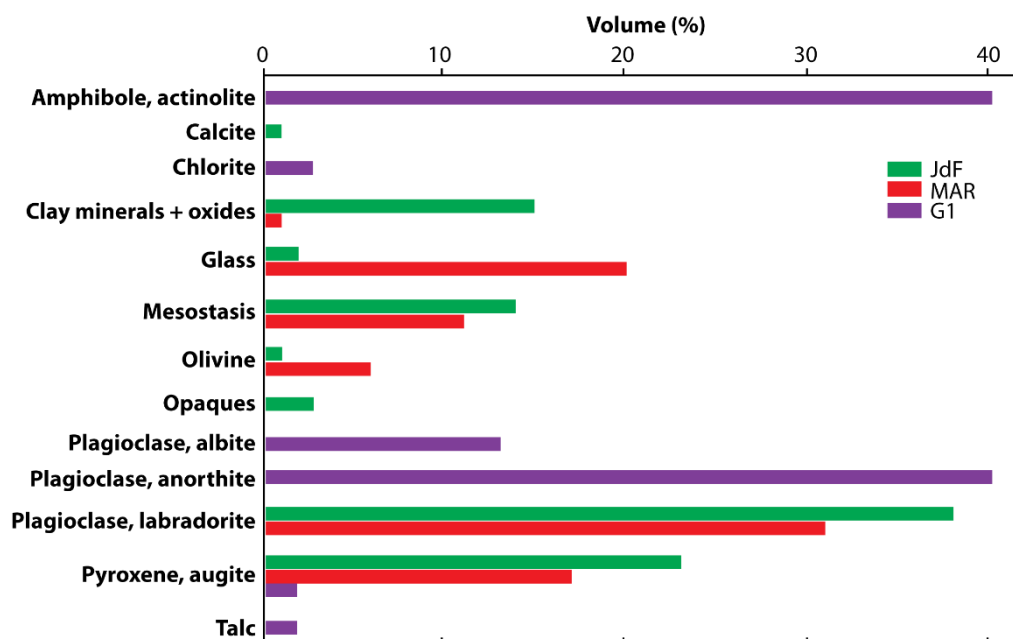


Figure 4.15: Mineral average distribution in the basalts of JdF and MAR, and in the ophiolitic gabbro G1.

The MAR and JdF basalts, despite the 15-30% of mesostasis and glass content, and ~20% of pyroxene, do not show particularly high dissolution rates (Table 4.12). However, the experimental run with the JdF samples “mix 1” and “mix 2”, and the ophiolitic gabbro show the highest aqueous concentration of Ca. This higher Ca-reactivity from the basalts is justified by the presence of ~1% of calcite, which is recognised in literature as very reactive [Cubillas *et al.*, 2005] (Table 4.15). In contrast, considering that G1 had no calcite, and almost no pyroxene (~2%), the main mineral sources of Ca in the gabbro have to be amphibole and anorthitic plagioclase.

No experimental data have been found in literature on single-phase Ca dissolution rates of amphiboles (Table 4.15), but Si results from whole rock dissolution experiments on metabasalt with 50% of actinolite content are shown by Critelli *et al.* [2014]. These whole rock Si dissolution rates are in agreement with this thesis findings (see further discussion in section 4.4.3). On the other hand, several experiments were carried out on plagioclase, even if only a small number measured the Ca dissolution rates (Table 4.15) [Berg and Banwart, 2000; Carroll and Knauss, 2005; Casey *et al.*, 1991]. Overall, these single-phase investigations demonstrate that the plagioclase dissolution rates change as a function of the plagioclase composition. The more calcic fraction (An-anorthite) exhibits faster reaction rates at a range of pH and temperatures [Blum, 1994; Casey *et al.*, 1991; Gudbrandsson *et al.*, 2014; Holdren and Speyer, 1987; Oelkers and Schott, 1995; Oelkers *et al.*, 1994], than the more sodic fractions, from labradorite to albite [Carroll and Knauss, 2005; Chou and Wollast, 1985; Gudbrandsson *et al.*, 2014; Oxburgh *et al.*, 1994; Stillings *et al.*, 1996] (Table 4.15). Hence, the fast Ca release in the experimental run with gabbro is in agreement with literature data on anorthitic plagioclase, but does not exclude a Ca contribution from the amphibole. Also, the slower Ca dissolution rates in this thesis basalts (JdF mix 3 and CD80WP132) can be related to the labradoritic nature of the plagioclase and/or to the lack of amphibole and calcite.

Table 4.15: Summary of mineral dissolution rates and relative experimental conditions at which they have been measured [Berg and Banwart, 2000; Carroll and Knauss, 2005; Casey et al., 1991; Chou and Wollast, 1985; Cubillas et al., 2005; Golubev et al., 2005; Gudbrandsson et al., 2014; Knauss et al., 1993; Oelkers and Schott, 1995; Oxburgh et al., 1994; Pokrovsky and Schott, 2000].

| Mineral                                                                                                                                | Reactor setup               | T °C   | pCO <sub>2</sub> bar | pH      | Si release rate log (mol/cm <sup>2</sup> /s) | pH      | Ca release rate log (mol/cm <sup>2</sup> /s) | Reference                   |
|----------------------------------------------------------------------------------------------------------------------------------------|-----------------------------|--------|----------------------|---------|----------------------------------------------|---------|----------------------------------------------|-----------------------------|
| Pyroxene diopside [Ca <sub>0.99</sub> Mg <sub>0.98</sub> Fe <sub>0.02</sub> Cr <sub>0.01</sub> Si <sub>2</sub> O <sub>6</sub> ]        | mixed flow <sup>a</sup>     | 25     | 0                    | 1-12    | from -14.12 to -15.32                        | 1-4     | from -14.04 to -14.75                        | Golubev et al. [2005]       |
| Pyroxene diopside [Ca <sub>0.97</sub> Mg <sub>0.89</sub> Fe <sub>0.08</sub> Al <sub>0.04</sub> Si <sub>1.99</sub> O <sub>6</sub> ]     | flow-through <sup>b</sup>   | 25-70  | atm                  | 4.25    | -14.62                                       | 4.25    | -14.56                                       |                             |
| Amphibole hornblende [Ca <sub>2</sub> (Mg,Fe,Al) <sub>5</sub> (Si,Al) <sub>8</sub> O <sub>22</sub> (OH) <sub>2</sub> ]                 | mixed flow <sup>a</sup>     | 25     | 0                    | 2-12    | from -12.80 to -15.27                        | 2-12    | from -12.54 to -15.83                        | Knauss et al. [1993]        |
|                                                                                                                                        |                             | 25     | 1                    | 2-11    | from -14.24 to -14.96                        | -       | -                                            | Golubev et al. [2005]       |
|                                                                                                                                        |                             |        |                      | 4       | -14.71                                       | -       | -                                            |                             |
| Forsterite [Mg <sub>2</sub> SiO <sub>4</sub> ]                                                                                         | mixed flow <sup>a</sup>     | 25     | 0                    | 2-11    | from -11.92 to -14.07                        | -       | -                                            | Golubev et al. [2005]       |
|                                                                                                                                        |                             | 25     | 1                    | 4.16    | -12.26                                       | -       | -                                            |                             |
| Forsterite [Mg <sub>2</sub> SiO <sub>4</sub> ]                                                                                         | mixed flow <sup>a</sup>     | 25     | -                    | 1-12    | from -11.23 to -15.25                        | -       | -                                            | Pokrovsky and Schott [2000] |
| Calcite [CaCO <sub>3</sub> ]                                                                                                           | mixed flow <sup>a</sup>     | 25     | -                    | -       | -                                            | 5.1-9.3 | from -9.57 to -10.35                         | Cubillas et al. [2005]      |
| Plagioclase Albite [Na <sub>0.97</sub> K <sub>0.01</sub> Ca <sub>0.02</sub> Al <sub>0.99</sub> Si <sub>3.00</sub> O <sub>8</sub> ]     |                             |        |                      |         | -14.97                                       | -       | -                                            |                             |
| Plagioclase Andesine [Na <sub>0.46</sub> K <sub>0.05</sub> Ca <sub>0.46</sub> Al <sub>1.50</sub> Si <sub>2.50</sub> O <sub>8</sub> ]   | batch <sup>c</sup>          | 25     | -                    | 2       | -14.16                                       | 2       | -14.07                                       | Casey et al. [1991]         |
| Plagioclase Anorthite [Na <sub>0.98</sub> Ca <sub>0.92</sub> Al <sub>1.90</sub> Si <sub>2.09</sub> O <sub>8</sub> ]                    |                             |        |                      |         | -12.19                                       | -       | -12.25                                       |                             |
| Plagioclase Albite [NaAlSi <sub>3</sub> O <sub>8</sub> ]                                                                               | fluidized bed <sup>d</sup>  | 25     | atm                  | 1-12    | from -14.2 to -15.9                          | -       | -                                            | Chou and Wollast [1985]     |
| Plagioclase Albite [Na <sub>0.99</sub> Ca <sub>0.01</sub> Al <sub>1.04</sub> Si <sub>2.97</sub> O <sub>8</sub> ]                       |                             |        |                      |         | from -13.42 to -15.33                        | -       | -                                            |                             |
| Plagioclase Labradorite [Na <sub>0.34</sub> Ca <sub>0.66</sub> Al <sub>1.65</sub> Si <sub>2.35</sub> O <sub>8</sub> ]                  | mixed flow <sup>a</sup>     | 22     | -                    | 2-11    | from -13.31 to -15.47                        | -       | -                                            | Gudbrandsson et al. [2014]  |
| Plagioclase Anorthite [Na <sub>0.13</sub> Ca <sub>0.87</sub> Al <sub>1.83</sub> Si <sub>2.14</sub> O <sub>8</sub> ]                    |                             |        |                      |         | from -12.05 to -14.62                        | -       | -                                            |                             |
| Plagioclase Anorthite [Ca <sub>1.2</sub> Al <sub>1.94</sub> Si <sub>2.0</sub> O <sub>8</sub> ]                                         | thin-film flow <sup>e</sup> | 25     | 0.1                  | 5.5-8.5 | from -15.06 to -15.36                        | 5.5-8.5 | from -15.66 to -15.76                        | Berg and Banwart [2000]     |
| Plagioclase Anorthite [Na <sub>0.04</sub> Ca <sub>0.97</sub> Al <sub>1.94</sub> Si <sub>2.04</sub> Fe <sub>0.02</sub> O <sub>8</sub> ] | mixed flow <sup>a</sup>     | 45-95  | -                    | 2.4-3.2 | from -10.55 to -11.52                        | -       | -                                            | Oelkers and Schott [1995]   |
| Plagioclase Labradorite [Na <sub>0.4</sub> Ca <sub>0.6</sub> Al <sub>1.6</sub> Si <sub>2.4</sub> O <sub>8</sub> ]                      | mixed flow <sup>a</sup>     | 30-130 | atm                  | 3.2     | from -11.66 to -14.57                        | 3.2     | from -11.69 to -13.40                        | Carroll and Knauss [2005]   |
| Plagioclase Oligoclase [Na <sub>0.87</sub> Ca <sub>0.13</sub> Al <sub>1.2</sub> Si <sub>2.8</sub> O <sub>8</sub> ]                     |                             |        |                      |         | from -12.34 to -13.45                        | -       | from -12.34 to -13.19                        |                             |
| Plagioclase Andesine [Na <sub>0.54</sub> Ca <sub>0.46</sub> Al <sub>1.4</sub> Si <sub>2.6</sub> O <sub>8</sub> ]                       |                             |        |                      |         | from -15.5 to -16.2                          | -       | -                                            |                             |
| Plagioclase Bytownite [Na <sub>0.24</sub> Ca <sub>0.76</sub> Al <sub>1.8</sub> Si <sub>2.2</sub> O <sub>8</sub> ]                      | flow-through <sup>b</sup>   | 25     | -                    | 3.0-6.5 | from -14.2 to -14.8                          | -       | -                                            | Oxburgh et al. [1994]       |
|                                                                                                                                        |                             |        |                      |         | from -13.3 to -14.5                          | -       | -                                            |                             |

<sup>a</sup> mixed flow: the solution flows into the reactor, is mixed with a stirrer, and leaves the reactor;

<sup>b</sup> flow-through: the solution passes continuously through the reactor, and gets progressively discharged;

<sup>c</sup> batch: the solution is present from the beginning in the reactor;

<sup>d</sup> fluidized bed: the solution passes continuously through a solid granular material at high velocities to suspend the solid, maximizing mixing and fluid-rock interaction;

<sup>e</sup> thin-film flow: the solution passes continuously through a thin membrane filter, where the powder is sitting on.

#### 4.4.3. Comparison with previous rock dissolution experiments

Seawater has a high ionic strength and its use in batch dissolution experiments, especially at low temperature, complicates the detection of relatively small contributions of cations to solution as a result of CO<sub>2</sub>-seawater-rock reactions. For this reason, at present, only a few experimental studies have investigated fluid-rock interactions in CO<sub>2</sub>-rich seawater [Crovisier *et al.*, 1987; Seyfried and Mottl, 1982; Wolff-Boenisch *et al.*, 2011]. Among them, data from Wolff-Boenisch *et al.* [2011] are comparable to the current study results, due to similarities in the experimental setup (Table 4.1). In that study, similar mixed flow dissolution experiments on mafic and ultramafic rocks (45-125 µm) were carried out at 25 °C and pCO<sub>2</sub> of 4 bar, continuously re-circulating the CO<sub>2</sub> in the system. The purpose of the study was to compare the steady-state silica release rates from basaltic glass (G), crystalline basalt (X) of similar chemical composition, as well as dunitic peridotite (P). The composition of these rocks, together with JdF, CD80WP132, and G1 samples from these study are summarised in Table 4.16, normalized to one mole of silicon.

Table 4.16: Composition comparison between basalts (JdF mix 1, JdF mix 2, JdF mix 3, and CD80WP132), and gabbro (G1) used in this study with the rocks used by Wolff-Boenisch *et al.* [2011], Critelli *et al.* [2014] and Gudbrandsson *et al.* [2011] (G-basaltic glass; X-crystalline basalt; P-peridotite; MB-metabasalt). All the chemical compositions have been normalized to one mole of silicon.

|                   | This study |           |           |           |        | Gudbrandsson <i>et al.</i> [2011],<br>Wolff-Boenisch <i>et al.</i> [2011] |       |       | Critelli <i>et al.</i><br>[2014] |
|-------------------|------------|-----------|-----------|-----------|--------|---------------------------------------------------------------------------|-------|-------|----------------------------------|
|                   | JdF mix 1  | JdF mix 2 | JdF mix 3 | CD80WP132 | G1     | G                                                                         | X     | P     | MB                               |
| Si                | 1.000      | 1.000     | 1.000     | 1.000     | 1.000  | 1.000                                                                     | 1.000 | 1.000 | 1.000                            |
| Ti                | 0.029      | 0.026     | 0.026     | 0.019     | 0.002  | 0.025                                                                     | 0.025 |       |                                  |
| Al                | 0.322      | 0.331     | 0.323     | 0.302     | 0.452  | 0.365                                                                     | 0.329 | 0.017 | 0.346                            |
| Fe <sub>tot</sub> | 0.178      | 0.173     | 0.175     | 0.222     | 0.093  | 0.194                                                                     | 0.193 | 0.120 | 0.233                            |
| Fe(II)            |            |           |           |           |        |                                                                           | 0.174 | 0.120 |                                  |
| Fe(III)           |            |           |           |           |        |                                                                           | 0.019 |       |                                  |
| Cr                | 0.001      | 0.001     | 0.001     | 0.0005    | 0.001  |                                                                           |       | 0.005 |                                  |
| Mn                | 0.004      | 0.003     | 0.003     | 0.004     | 0.002  | 0.003                                                                     | 0.003 | 0.002 |                                  |
| Mg                | 0.195      | 0.227     | 0.202     | 0.233     | 0.368  | 0.294                                                                     | 0.310 | 1.639 | 0.472                            |
| Ca                | 0.268      | 0.306     | 0.275     | 0.282     | 0.408  | 0.263                                                                     | 0.273 |       | 0.272                            |
| Na                | 0.092      | 0.086     | 0.082     | 0.067     | 0.041  | 0.081                                                                     | 0.061 |       | 0.063                            |
| K                 | 0.005      | 0.003     | 0.005     | 0.001     | 0.001  | 0.008                                                                     | 0.007 |       | 0.002                            |
| P                 | 0.003      | 0.001     | 0.001     | 0.001     | <0.001 | 0.004                                                                     | 0.003 |       |                                  |
| S                 | 0.002      | 0.002     | 0.001     | 0.004     | 0.001  |                                                                           |       |       |                                  |
| O                 | 3.435      | 3.468     | 3.370     | 3.388     | 3.865  | 3.403                                                                     | 3.374 | 3.795 | 3.727                            |

Among the several experimental series performed by *Wolff Boenisch et al.*, the “SWC” focussed on the effect of the chemical components of seawater on dissolution rates. The initial values of this series, called “SWC-CI”, represent the steady-state composition of the solution, making them the most suitable data for comparison. The Si dissolution rates as a function of pH are summarised in Figure 4.16, together with the elemental rates from all the experimental runs of this study. The Si release rates in *Wolff-Boenisch et al.* were generated using:

$$r_{Si} = \log [F * C_{Si} / (A_{geo_i} * m)] \quad (15)$$

where  $r_{Si}$  is the dissolution rate of Si expressed as  $\log [\text{mol}/\text{cm}^2/\text{s}]$ ,  $F$  is the pump flowrate equal to 2 mL/min,  $C_{Si}$  is the concentration of Si in the outlet fluids in mol/L,  $A_{geo}$  the initial geometric surface area of sample  $i$  in  $\text{cm}^2/\text{g}$ , and  $m$  the initial mass of material in the reactor in g. No grain size distribution analysis was performed on the ground materials, and no Ca dissolution rates were measured.

The mafic and ultramafic rocks from *Wolff-Boenisch et al.* exhibit significantly higher rates than oceanic rocks in this study, even if the major cation chemistry is relatively similar (Table 4.16). The explanation can be related to three crucial factors:  $\text{pCO}_2$ , water/rock ratio, and experimental setup itself (Table 4.17).

Table 4.17: Summary of experimental conditions of current study and previous works.

|                                            | Reactor setup        | T<br>°C | pCO <sub>2</sub><br>bar | DIC<br>mM | pH     | Water/rock<br>ratio<br>mL/g |
|--------------------------------------------|----------------------|---------|-------------------------|-----------|--------|-----------------------------|
| <b><i>This study</i></b>                   | batch, seawater      | 40      | 1                       | <40       | 4.8    | 6-17                        |
| <b><i>Critelli et al. (2014)</i></b>       | mixed flow           | 25      | -                       | -         | 2 - 12 | 67-300                      |
| <b><i>Wolff-Boenisch et al. (2011)</i></b> | mixed flow, seawater | 25      | 4                       | 114       | 3.5    | 60-75                       |
| <b><i>Gislason and Oelkers (2003)</i></b>  | mixed flow           | 25 - 50 | -                       | -         | 2 - 12 | n.a.                        |
| <b><i>Gudbrandsson et al. (2011)</i></b>   | mixed flow           | 5 - 75  | -                       | -         | 2 - 11 | 60-300                      |

In the case of mixed-flow experiments the 4 bar  $p\text{CO}_2$  led to ~114 mM of dissolved inorganic carbon in solution, which is two orders of magnitude higher than in these new batch experiments, resulting in a much lower pH of 3.5 (Table 4.16). The lower the pH, the higher the activity of  $\text{H}^+$ , and the higher the dissolution rate [Gislason and Oelkers, 2003; Wolff-Boenisch et al., 2004]. Also, Golubev et al. [2005] found that silicate dissolution rates can be affected by the presence of higher  $\text{CO}_2$  in solution, independently from the pH.

The second factor potentially responsible for the rate discrepancies, the water/rock ratio, was ~60-75 mL/g in the mixed-flow reactors, whereas ~6-17 mL/g in the HDPE bottles of this study, where only the top-surface always in contact with seawater was most probably reactive (Section 4.4.2.2). This means that the higher the water/rock ratio is, the higher the solid surface in contact with the reactive solution is, potentially resulting in a higher dissolution rate. This highlights the major role of the reactive mass quantification in the understanding of the extent and rate of rock dissolution.

Finally, the importance of the experimental setup in the measurement of dissolution rates has been observed by many authors [Clow and Drever, 1996; van Grinsven and van Riemsdijk, 1992], who measured higher rates with stirred (“mixed”) batch reactors than with column experiments (“flow-through”) or non-stirred batch reactors.

The dissolution rate dependency on general experimental settings is further confirmed by Gislason and Oelkers [2003] for basaltic glass mixed flow experiments at 25 and 50 °C as a function of pH, in an aqueous  $\text{CO}_2$ -free solution with lower ionic strength than seawater (0.1 and 0.7 mol/kg, respectively) (Table 4.17, and Figure 4.16). The 25 and 50 °C elemental dissolution curves, which were obtained by a multioxide dissolution model, are closer to the dissolution rates of the samples used in this batch reactor study, and show slower rates than Wolff-Boenisch et al., despite the common mixed flow setup (Figure 4.16).

Hence, all the previously listed factors are potentially affecting the dissolution rate, but this discrepancy seems to be mainly driven by the pCO<sub>2</sub> conditions.

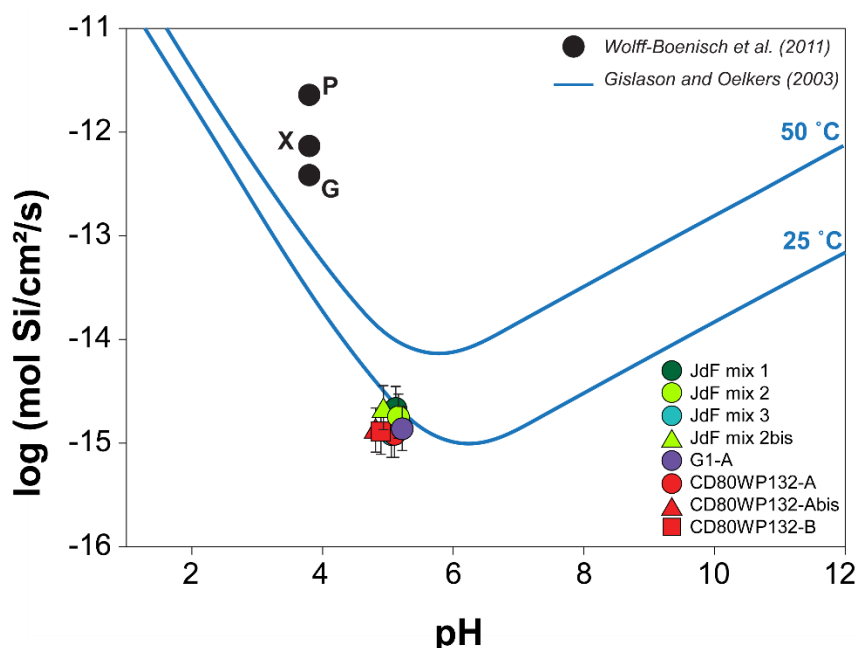


Figure 4.16: Si release rates “surface corrected” as a function of pH, for all the experimental runs of this study at 40 °C (coloured symbols) and for SWC-CI series from Wolff-Boenisch et al. [2011], where G-basaltic glass, X-crystalline basalt, and P-peridotite. The blue lines indicate the overall dissolution rates proposed by Gislason and Oelkers [2003] for basaltic glass mixed-flow experiments at 25 and 50 °C as a function of pH, based on multioxide dissolution model. Error bars correspond to a  $\pm 0.2$  log unit estimated uncertainty calculated from the precision of Si concentrations measurements (Appendix “Methods” CH4-2, IAPSO error).

The whole rock Si release rates “surface corrected” from this study can also be compared with the elemental release rates obtained by Critelli et al. [2014], and Gudbrandsson et al. [2011], following Equation 15 (Table 4.2, and Figure 4.17). The authors carried out mixed-flow dissolution experiments on a metabasalt from an ophiolitic sequence (Monte Reventino, Italy), and a crystalline basalt from the Stapafell Mountain (SW-Iceland), respectively, at pH from 2 to 12, temperatures from 5 to 75 °C, and water/rock ratio of ~60-300 mL/g, to investigate the steady state elemental release rates in a CO<sub>2</sub>-free aqueous solution (Table 4.17). The rock mineralogy is summarised in Table 4.18, whereas the XRF bulk analyses, normalized to one silicon, are shown in Table 4.16. The crystalline basalt used by Gudbrandsson et al. is the same as the X-crystalline basalt described by Wolff-Boenisch et al.



Table 4.18: Summary of rock mineralogy for metabasalt [Critelli et al., 2014], and crystalline basalt [Gudbrandsson et al., 2014] used during dissolution experiments. The relative mineral content is given in wt% and vol%, respectively.

|                          | <i>Critelli et al. [2014]</i> | <i>Gudbrandsson et al. [2011]</i> |
|--------------------------|-------------------------------|-----------------------------------|
| Amphibole, actinolite    | 50.0                          | -                                 |
| Calcite                  | 1.0                           | -                                 |
| Epidote                  | 9.0                           | -                                 |
| Glass                    | -                             | 4.2                               |
| Iron oxides              | -                             | 4.7                               |
| Olivine                  | -                             | 15.8                              |
| Phyllosilicate, chlorite | 32.0                          | -                                 |
| Phyllosilicate, phengite | 1.0                           | -                                 |
| Plagioclase, albite      | 7.0                           | -                                 |
| Plagioclase, labradorite | -                             | 41.3                              |
| Pyroxene                 | -                             | 34.0                              |

Figure 4.17 shows how the Si release rates from this study are broadly in agreement with the elemental rates measured at 25 °C by *Critelli et al. [2014]*, and *Gudbrandsson et al. [2011]* for pH ~5, lying across the labradorite [*Gudbrandsson et al., 2011; Oxburgh et al., 1994*] and hornblende [*Golubev et al., 2005*] dissolution curves estimated for 25 °C. These hornblende dissolution rates proposed by *Golubev et al.* are further confirmed by the actinolitic values computed by *Critelli et al.* from the whole rock rates.

Furthermore, *Gudbrandsson et al.* also proposed whole rock Ca dissolution rates for a range of pH (Figure 4.18). All the Ca release rates investigated in this thesis seem to be slightly faster than X-crystalline basalt, especially for the gabbro and the two JdF samples “mix 1” and “mix 2” (Figure 4.18). This rate difference is most probably related to the mineralogy (Figure 4.18). In general, this comparison confirms the preferential dissolution of Ca at pH ~5 in all the rocks investigated during the several experimental runs, and suggests a major role of calcite, and plagioclase and/or amphibole composition in Ca dissolution rates.

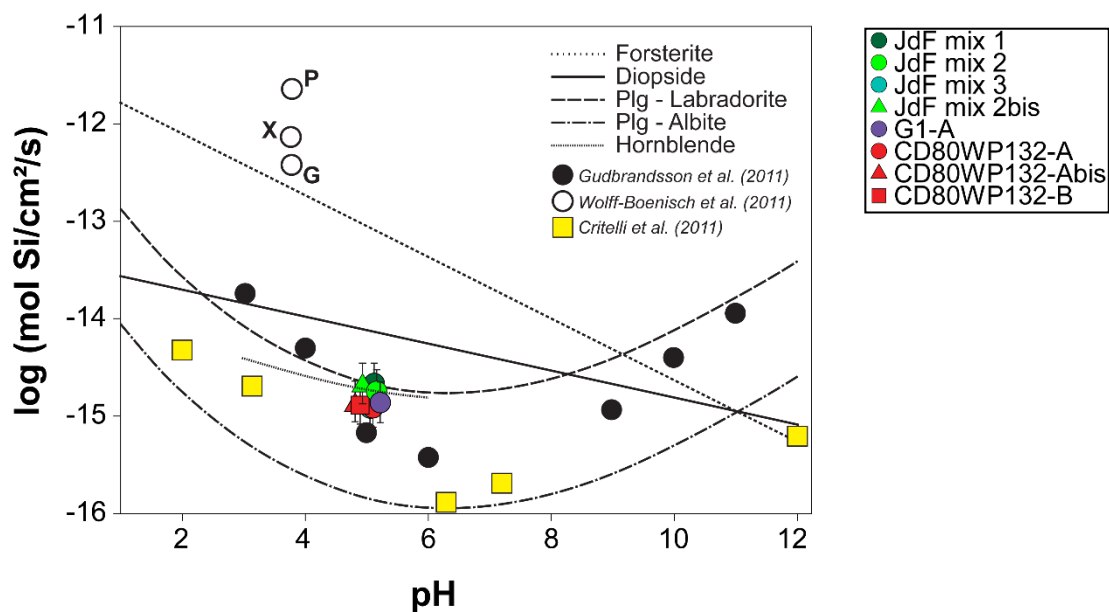


Figure 4.17: Whole rock Si dissolution rates “surface corrected” for all the experimental runs as a function of pH (coloured symbols), compared to experimental results of Critelli et al. [2014], Gudbrandsson et al. [2011] at 25 °C, and Wolff-Boenisch et al. [2011]. The forsterite, diopside, albite, and hornblende rates at 25 °C illustrated in this figure are taken from Pokrovsky and Schott [2000], Knauss et al. [1993], Chou and Wollast [1985], and Golubev et al. [2005], respectively. The labradorite curve was taken from Gudbrandsson et al. [2011], who in turn obtained it multiplying the albite dissolution curve of Chou and Wollast [1985] by 4 to fit the bytownite and andesine rate data taken from Oxburgh et al. [1994]. Error bars correspond to a  $\pm 0.2$  log unit estimated uncertainty calculated from the precision of Si concentrations measurements (Appendix “Methods” CH4-2, IAPSO error).

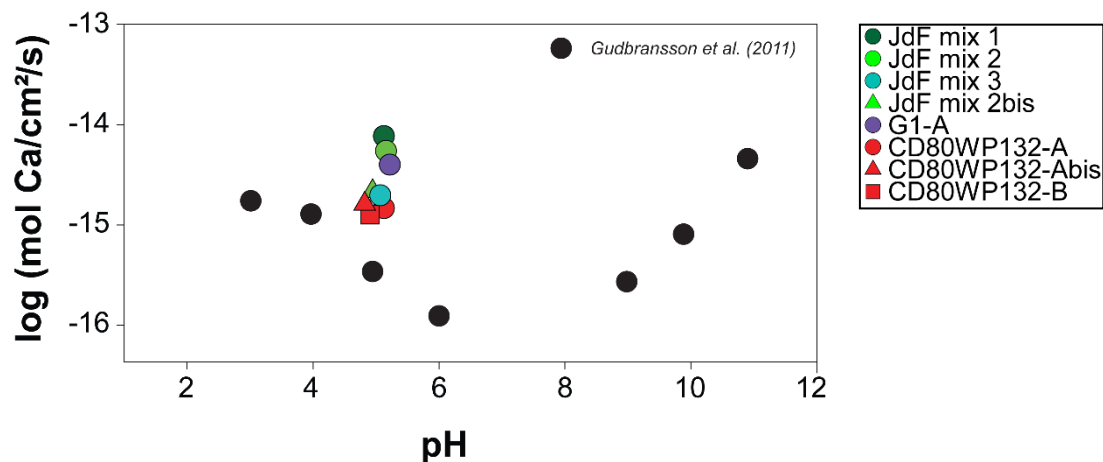


Figure 4.18: Whole rock Ca dissolution rates “surface corrected” for all the experimental runs as a function of pH (coloured symbols), compared to the experimental results of Gudbrandsson et al. [2011] at 25 °C. Size of symbols correspond to a  $\pm 0.01$  log unit estimated uncertainty calculated from the precision of Ca concentrations measurements (Appendix “Methods” CH4-2, IAPSO error).

It should be noted that the Si dissolution rates measured by Gudbrandsson et al. [2011] (from -14 to -15 log (mol/cm²/s)) are significantly lower than the one presented by of

Wolff-Boenisch *et al.* [2011] ( $-12 \log (\text{mol}/\text{cm}^2/\text{s})$ ) (Table 4.2, Figure 4.17), even though they both conducted mixed flow experiments on crystalline rocks, using the same apparatus in the same laboratory, and a similar water-rock ratio (Table 4.17, and Figure 4.17). This means that, considering outputs at the same pH (i.e. pH ~3), the main factors potentially influencing this rate difference are  $\text{pCO}_2$  [Golubev *et al.*, 2005], and the solution composition (Table 4.1). In fact, as described in Section 4.1, the behaviour of the NaCl contained in seawater, as well as a number of acids and bases in solution (such as  $\text{H}_2\text{SO}_4$ ,  $\text{HNO}_3$ ,  $\text{F}^-$ ,  $\text{NaOH}$ , and  $\text{NaHCO}_3$ ), and their catalytic effect on the Si dissolution rates, are described in a number of studies [Gadikota *et al.*, 2014a; Kaszuba *et al.*, 2005; Oelkers, 2001b; Teir *et al.*, 2007; Wolff-Boenisch, 2011; Wolff-Boenisch *et al.*, 2011].

The different experimental setup of Gudbrandsson *et al.* [2011] and Critelli *et al.* [2014] in comparison to this study (mixed flow and batch reactors, respectively) does not seem to significantly influence the experimental outcome (Figures 4.17 and 4.18).

Hence, further investigation on Si dissolution rates of JdF, CD80WP132, and G1 should be carried out with a variety of experimental setups, such as flow-through or stirred mixed-flow, varying  $\text{pCO}_2$  values, and water/rock ratios, to have an improved perspective on their reactivity and to define the dependence on each experimental factor.

#### 4.5. Conclusions

Geochemical trapping of CO<sub>2</sub> is part of a Carbon Capture and Storage mitigation strategy to reduce the CO<sub>2</sub> concentration in the atmosphere. This mechanism has been found to be more efficient (time required <1000 yr) in the presence of mafic and ultramafic formations, due to their fast cation release rates.

The aim of this chapter was to determine the dissolution rate in seawater-rock reaction experiments for oceanic rocks at low T (40°C), and pCO<sub>2</sub> of 1 bar. The main findings are:

- the measurement of grain size distribution in the samples is essential for accurate interpretation of experimental results;
- the experimental setup adopted in this study lead to solution evaporation during the experimental run, resulting in a mass loss from 1 to 20% in each reactor. Therefore the evaporation extent has been measured using Cl, SO<sub>4</sub><sup>2-</sup> concentrations, assuming these elements are conserved, and applied for correction to the resulting cation concentrations in solution;
- Si and Ca are the key elements to detect the release of cations from the solid to the seawater solution; other dissolved ions (e.g. Na, Mg, and K) do not show consistent chemical variations in the solutions with time due to their high concentrations in seawater;
- congruent dissolution was not reached most probably because of the formation of a passivation layer, enriched in Si and depleted in metals (e.g. Ca);
- incongruent dissolution during most of the elapsed time is due to reactions focussed on the surface of mineral grains;
- the mass balance calculations on Si and Ca concentrations suggest the presence of surface reaction mechanisms, showing that the top-surface only, with less than 1 g of rock involved in the reactions (assuming a full Si and Ca dissolution of grains with SiO=50 wt% and CaO = 12 wt%), could provide all the dissolved ions;

- at pH ~5, whole rock dissolution rates for Si and Ca range from -14.67 to -14.92 log (mol/cm<sup>2</sup>/s), and from -14.10 to -14.88 log (mol/cm<sup>2</sup>/s), respectively, and represents the closest rates to the stoichiometric dissolution;
- the similarity of Si release rate of ophiolitic gabbro and basalts, and the higher Ca release rate from the gabbro, JdF mix 1, and JdF mix 2 is perhaps one of the most significant observation of this study;
- the interpretation of Si and Ca dissolution rates identifies calcite, together with amphibole and plagioclase as main calcium sources in these batch dissolution experiments;
- overall, the crystalline samples used in this study show a similar or higher reactivity to Icelandic crystalline basalt [*Gudbrandsson et al.*, 2011], and metabasalt [*Critelli et al.*, 2014], both reacted in a mixed flow reactor at 25 °C, but lower than basaltic glass investigated by *Gislason and Oelkers* [2003], and *Wolff-Boenisch et al.* [2011];
- solution composition, and pCO<sub>2</sub> conditions seem to affect the dissolution rate outputs from experiments carried out at similar temperature and pH, independently from the experimental setup;
- further investigations on JdF, CD80WP132, and G1 should be carried out with different experimental setups, such as flow-through or stirred mixed-flow, at varying solution compositions, pCO<sub>2</sub> and water/rock ratio conditions, to develop a more complete perspective on their reactivity and to define the extent of each experimental factors.



## Chapter 5: Costs linked with offshore CO<sub>2</sub> sequestration projects in oceanic crust

### 5.1. Introduction

In the last decades, the strategy of storing carbon dioxide in geological formations to reduce anthropogenic CO<sub>2</sub> emissions to the atmosphere has started to be implemented, but is still developing scientifically and technologically [IPCC, 2005b]. To date, carbon storage is mainly undertaken in sedimentary formations, such as deep aquifer and depleted oil and gas fields – EOR [Global CCS Institute, 2015]. For instance, injection of CO<sub>2</sub> into the subsurface has been practiced for decades for EOR [Al Eidan et al., 2015]. The cost of carbon storage is directly related to type of capture and transport facility, storage capacity, and reservoir utilisation [Leung et al., 2014]. Among the studies and reports focused on the costs of CO<sub>2</sub> transport and storage in a CCS context (e.g. *Global CCS Institute* [2015]; *IEA* [2015]; *Rubin* [2008]), a growing number of articles consider offshore storage options (e.g. *Aspelund* [2010]; *Decarre et al.* [2010]; *Haugen et al.* [2009]; *McKinsey & Company* [2008]; *Rubin et al.* [2015]; *Svensson et al.* [2005]; *Torp and Brown* [2004]). However, in most studies, the scenarios and explanations are not sufficiently detailed to enable direct comparisons of proposed costs. The most comprehensive study on CCS costs has been carried out by the European Technology Platform for Zero Emission Fossil Fuel Power Plants (known as the ZEP – Zero Emissions Platform), which represents a coalition of stakeholders supportive of CCS [ZEP, 2011a].

Despite the potential advantages of carbon sequestration in mafic formations (Chapters 2, 3, and 4), the exploitation of oceanic crust for CCS purposes has not been considered because of the supposed high costs involved with offshore reservoirs. To the author's

knowledge, there have been no publications to date that breakdown the expenditures associated with a CCS project in its entirety to allow the assessment of the economic feasibility of offshore strategies, leaving a number of questions unsolved. For example:

- What is the cost of offshore CO<sub>2</sub> sequestration?
- Specifically, what is the cost of offshore transport and storage of CO<sub>2</sub> in deep-sea basalts?
- What infrastructures are required?
- What are the main limitations related to offshore injection of CO<sub>2</sub>?
- Are there economically feasible offshore locations for CCS projects in deep-sea basalts?

In this chapter, an attempt is made to address these questions, estimating the cost linked with offshore annual transport and storage of 20 Mt of CO<sub>2</sub> in the oceanic crust. The costs are considered for distances of 500, 1000, and 1500 km from the coast, and typical well injection rates of 0.5 to 3 Mt CO<sub>2</sub>/yr. Finally, the overall costs related to a potential offshore CCS project in deep-sea basalts have been calculated for three scenarios, on the basis of water depth, with potential reservoirs located at 1000, 2500, and 5000 mbsl. Costs are giving in euros (€), and at the time of writing the exchange rate with US dollars is €1.00 = \$1.13 [<http://www.xe.com/currencyconverter/convert/?From=EUR&To=USD>].

## **5.2. Costs for CO<sub>2</sub> transport offshore**

Three distance-specific cases of CO<sub>2</sub> offshore transport are presented, with CO<sub>2</sub> costs based on ZEP estimations [ZEP, 2011a] and updated by Mr. Nils H. Eldrup (project manager at Tel-Tek and member of ZEP-transport working group). The ZEP report provides cost estimates for large-scale CCS, including costs for different pipes and ships for offshore networks.



Carbon dioxide from two point sources of 10 Mt/yr each (equal to  $20 \times 10^6$  t/yr) is transported by ~10 km onshore feeders to a coastline port from where it is shipped to storage sites located 500, 1000, or 1500 km offshore (Figure 5.1). For example: a typical coal plant with a capacity of 1000 MW, is able to power up to 1 million houses, and produces ~6 Mt of CO<sub>2</sub> per year. Consequently, the shipping of 20 Mt/yr of CO<sub>2</sub> corresponds to the cumulative annual emissions of at least three of these power stations.

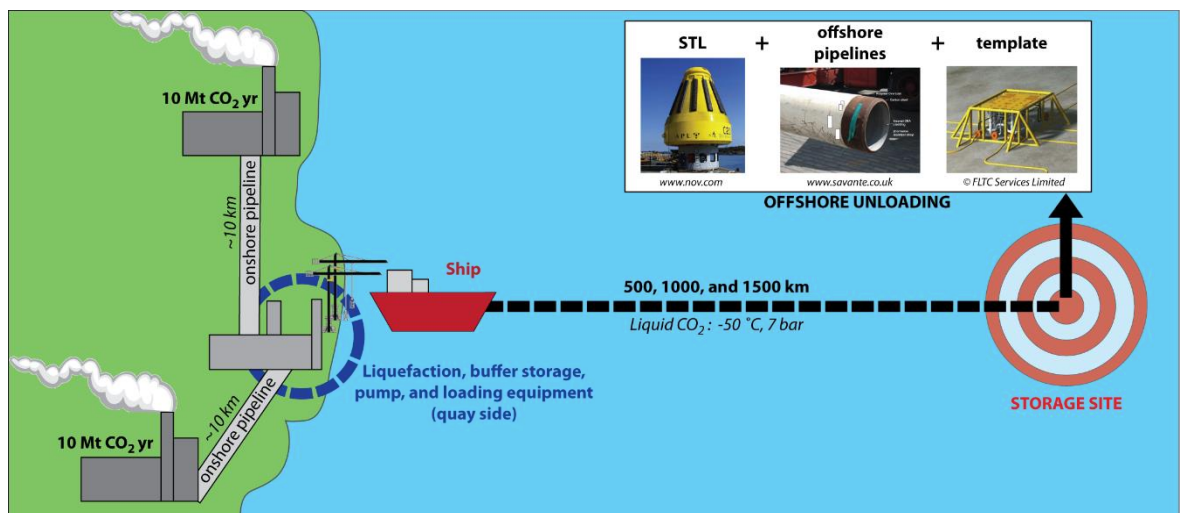


Figure 5.1: Network of CO<sub>2</sub> offshore transport via ship, from two point-sources ( $2 \times 10$  Mt CO<sub>2</sub>/yr) along the coastline to a storage site at 500, 1000, or 1500 km from shore. STL = Submerged Turret Loading.

Ship transport costs have been estimated following the assumptions listed below [ZEP, 2011a]:

- the loading equipment and functional quay facilities are in existence, and the loading time is set at 12 h;
- the liquefaction plant on the coast delivers CO<sub>2</sub> at 7 bar (0.7 MPa) and -50 °C to the storage tanks; these pressure and temperature conditions have been chosen to meet the best cost/requirement ratio [Svensson *et al.*, 2005], and to avoid any risk of formation of dry ice;
- the transport process is assumed to deliver the CO<sub>2</sub> to the well-head at the storage site in the following condition:
  - temperature = ambient seawater temperature from 4 to 15 °C;

➤ pressure = 60 barg (6 MPa) - barg, or “*gauge pressure*”, is equal to absolute pressure minus atmospheric pressure;

- the maximum cargo capacity per ship is equal to 40,000 m<sup>3</sup> (~1x10<sup>9</sup> mol of CO<sub>2</sub>);
- the average ship speed is 12 knots (~22 km/h);
- no intermediate offshore storage is assumed to be necessary and results in an intermittent injection into the wells with a discharge time set at 48 h;
- CAPEX (capital expenditure) is calculated for a project lifetime of 25 years with an interest rate of 8%, based on the weighted average cost of capital (WACC);
- OPEX (annual operational expenditure) is based on actual operating cost experience: crew costs, maintenance, fuel costs, and port fees;
- Cost estimate accuracy = ±30%.

In addition to the cost of transport distance from shore, ship transport also includes the costs of liquefaction, a storage ship with dynamic positioning and unloading buoy (*STL-Submerged Tool Loading*<sup>™</sup>), and flexible pipelines to the seafloor limited to a first *template* (supporting structure on the seabed). Details on the transport facilities and relative transport costs are summarised in Tables 5.1 and 5.2, respectively.

The *STL system* consists of a conventional cargo tanker which connects/disconnects to a submerged cone-shaped turret buoy, which in turn is catenary moored to pile anchors [Wall *et al.*, 2002]. The mooring buoy can be simply disconnected from the transport ship, where it would float in an equilibrium position about 30 – 50 m below the sea surface. A vessel equipped with *STL* tools is capable of staying moored to the transfer system at a location offshore and perform its function even in severe weather conditions. In general, 3 to 4 buoys have been considered in the estimations for all the scenarios, but this number may be greater for larger scale projects, because it is a function of the number of wells and ships involved.

Both the *STL system* and offshore pipelines have been designed to be capable of holding at 250 barg, which corresponds to water depth of ~2500 m.

The *template* is a steel protection structure with integrated manifolds and wellheads, and is considered to be a four-slot subsea terminal with a 50 km control cable to a platform or an on-site vessel. A typical example of a vessel permanently moored to the seabed is the FPSO for LNG: Floating Production Storage and Offloading system for Liquefied Natural Gas. The LNG FPSO is a floating facility usually installed above or close to an offshore gas field in order to receive, process, liquefy, store and export natural gas. The first LNG platform, as well as the largest offshore facility (488 m long, 74 m wide) ever constructed is the “Prelude FLNG (Floating Liquefied Natural Gas)”, located in the Browse Basin (NW Australia). At full load, it displaces more than 600,000 t (0.6 Mt), and cost ~€9 billion.

The costs for a platform or vessel, beside the cargo ships, are not taken into account in these estimations. The costs of manifolds for wells, drilling of standard injection wells, and connecting offshore pipelines are assumed to be part of the storage costs (Section 5.3, Table 5.3) [ZEP, 2011b].

Table 5.1: Details of the facilities and timings for which the cost estimations are based.

| Distance               | km    | 500 km | 1000 km | 1500 km |
|------------------------|-------|--------|---------|---------|
| Ship                   | n°    | 7      | 9       | 12      |
| Roundtrip per ship     | n°/yr | 70     | 50      | 39      |
| Roundtrip time         | days  | 5      | 7       | 9       |
| Sailing time (one way) | h     | 22     | 45      | 67      |
| STL Buoy               | n°    | 4      | 3       | 3       |

Following the ZEP report assumptions, at ship transport conditions (-50 °C, 7 bar) CO<sub>2</sub> density is equal to 1155 kg/m<sup>3</sup>, and the potential mass of transportable CO<sub>2</sub> for a ship with capacity of 40,000 m<sup>3</sup> is 0.046 Mt per journey. This means that a total of 435 discharges of full cargo ships are needed per year to achieve the target of 20 Mt of total

annual injectivity. A fleet from 7 to 12 ships, together with an appropriate number of STL buoys, is required, depending on the reservoir offshore distance, to deliver the expected 20 Mt of CO<sub>2</sub> at the storage site (Table 5.1). Also, considering 48 h as the assumed constant discharge time, and one template, the injection rate in each of the four wells is ~240 t/h.

The transport of 20 Mt of CO<sub>2</sub> per year via ship at 500 km from the shore (Table 5.2) can be estimated using the following formulas (Equations 1 and 2):

$$\text{Capital 1st year} = CAPEX_{f+s} + OPEX_{f+s} + annuity \quad (16)$$

$$\text{Capital following years} = OPEX_{f+s} + annuity \quad (17)$$

where  $CAPEX_{f+s}$  are the CAPEX costs for onshore feeders and ship transport [M€],  $OPEX_{f+s}$  the OPEX costs for onshore feeders and ship transport [M€/yr], *annuity* the fixed sum to be paid per year [M€/yr]. This type of transport requires at least ~€1233 M (€1233x10<sup>6</sup>) during the first year (Equation 1), and ~€186 M for each of the following years (Equation 2). The correspondent transport costs for 1500 km as offshore distance are estimated to be ~€1845 M during the first year, and ~€254 M for each of the following years (Table 5.2). Hence, the transport of 20 Mt of CO<sub>2</sub> per year, for the whole project lifetime of 25 years, ranges from ~€5.70 to ~€7.95 B (Table 5.2).

Table 5.2: Cost estimates for CO<sub>2</sub> offshore transport via ships, from two point-sources (2 \* 10 Mt CO<sub>2</sub>/yr) along the coastline to a storage site located 500, 1000, or 1500 km from shore.

| Distance km                 |            | 500         |               |                  | 1000        |               |                  | 1500        |               |                  |
|-----------------------------|------------|-------------|---------------|------------------|-------------|---------------|------------------|-------------|---------------|------------------|
|                             |            | CAPEX<br>M€ | OPEX<br>M€/yr | Annuity<br>M€/yr | CAPEX<br>M€ | OPEX<br>M€/yr | Annuity<br>M€/yr | CAPEX<br>M€ | OPEX<br>M€/yr | Annuity<br>M€/yr |
| Feeders onshore             | 2*10 Mt/yr | 30.1        | 0.1           |                  | 30.1        | 0.1           |                  | 30.1        | 0.1           |                  |
| Ship transport              |            | 860.0       | 156.3         | 29.7             | 1071.0      | 178.9         | 38.1             | 1355.0      | 204.9         | 49.4             |
| TOTAL                       | M€/yr      | 890.1       | 156.4         |                  | 1101.1      | 179.1         |                  | 1385.1      | 205.0         |                  |
| TOTAL for 20 Mt/yr          | M€/yr      | 5700        |               |                  | 6710        |               |                  | 7950        |               |                  |
| TOTAL CO <sub>2</sub> costs | €/t        | 11.4        |               |                  | 13.4        |               |                  | 15.9        |               |                  |

### 5.3. Costs for CO<sub>2</sub> storage offshore

The storage cost estimations [Eldrup, personal communication, February 2016] (Table 5.3) take into account manifolds and corresponding templates for wells, connecting offshore pipelines, drilling and casing of injection wells.

Several scenarios, based on the more typical well injectivity ranging from 0.5 to 3 Mt/yr, are presented to estimate the costs of the required offshore storage facilities for a reservoir able to store 20 Mt of CO<sub>2</sub> per year.

The well properties and drilling costs are those for cased holes drilled into sedimentary formations, with depth up to 3000 mbsf, which are characteristic of hydrocarbon production fields [ZEP, 2011b].

Table 5.3: Cost estimates for four offshore scenarios, each able to store 20 Mt CO<sub>2</sub> per year, taking into account injection rate, number of wells, pipeline distance and template.

| Annual injection rate | Injection rate | Number of wells | Well cost | Pipeline distance | Pipeline cost | Pipeline end module cost | Template cost | Total CAPEX cost |
|-----------------------|----------------|-----------------|-----------|-------------------|---------------|--------------------------|---------------|------------------|
| Mt/yr                 | t/h            |                 | M€        | km                | M€            | M€                       | M€            | M€               |
| 0.5                   | 57             | 40              | 2000      | 276               | 414           | 380                      | 1120          | 3914             |
| 1                     | 114            | 20              | 1000      | 136               | 204           | 190                      | 560           | 1954             |
| 2                     | 228            | 10              | 500       | 60                | 90            | 76                       | 280           | 946              |
| 3                     | 342            | 7               | 350       | 36                | 54            | 38                       | 196           | 638              |

If the annual well injectivity increases from 0.5 to 3 Mt CO<sub>2</sub>, the injection rate must increase from 57 to 342 t/h, resulting in a decrease in the number of injection wells required. This approach is due to the reservoir pressure, which is one of the most limiting factors for large scale geological storage and directly depends on the injection rate, formation permeability, thickness of formation, viscosity and compressibility of CO<sub>2</sub> [Maroto-Valer, 2010; Mathias et al., 2009]. The bottom-hole pressure should exceed the formation pressure to avoid the influx of *in situ* fluids into the wellbore [Bachu and Gunter,

2004]. However, to meet the safety criteria for prevention of hydraulic fracturing of the reservoir, regulations stipulate that the bottom-hole pressure should not exceed 90% of the fracturing pressure [IEAGHG, 2010], which is defined by *Hubbert and Willis* [1972] as (Equation 3):

$$P_{fr} = \sigma_{min} + P_f \quad (18)$$

where  $P_{fr}$  is the fracturing pressure,  $\sigma_{min}$  the minimum principal stress, and  $P_f$  the formation pressure (hydrostatically calculated). The fracturing pressure is the pressure at which the rock breaks, and the variation of pressure required to induce a fracture with depth is called the fracture gradient. An increase in the fracture gradient (i.e. at greater depth) corresponds to an increase in the maximum injection pressure. Consequently, in the case of a field with high reservoir pressure, it is viable to inject more CO<sub>2</sub> (high injection rate of 342 t/h), and the number of wells needed is reduced, along with the relative costs [IEAGHG, 2010].

In order to reach the target of 20 Mt of CO<sub>2</sub> stored annually in offshore reservoirs, the number of wells required varies from 40 to 7, with relative injection rate of 57 to 342 t/h, respectively. However, at constant discharge time of 48 h (Section 5.2), the number of wells needed to accommodate the content of one ship cargo (0.046 Mt CO<sub>2</sub>) varies from 17 to 3, respectively, depending on the injection rate.

The CAPEX for the storage of 20 Mt of CO<sub>2</sub> per year in 40 available injection wells, each with injectivity of 0.5 Mt CO<sub>2</sub>/yr, is estimated to be at least ~€3914 M, whereas for an higher injectivity of 3 Mt CO<sub>2</sub>/yr in 7 available injection wells it would be ~€638 M (Table 5.3). Although the last option offers a more cost-effective injection scenario, it will require reservoirs with particularly good hydraulic and physical properties to keep the site suitable for CO<sub>2</sub> storage.

#### 5.4. Summary of costs for CO<sub>2</sub> transport and storage offshore

Theoretically, if the total costs associated with the transport of 20 Mt/yr of CO<sub>2</sub> at reservoirs located 500 km from the shore, are added to the storage costs of seven wells with 3 Mt CO<sub>2</sub>/yr annual injection rate each, ~€1.7 B would be necessary during the first year. If the transport distance increases by 3 times to 1500 km, the first year required capital does not triple, but increases to ~€2.3 B. If the storage site is at 500 km offshore and the costs are related to forty wells, each with 0.5 Mt CO<sub>2</sub>/yr as annual injection rate, ~€5.0 B would be necessary during the first year.

This means that the cheapest scenario for both transport and storage cost estimations includes a reservoir located 500 km from the shore, comprising seven injection wells each with 3 Mt CO<sub>2</sub>/yr injection rate, and the 20 Mt of CO<sub>2</sub> per year delivered by a fleet of 7 ships (Table 5.4). On the other hand, the most expensive options involve a storage site at 1500 km from the coastline, with 40 wells that have 0.5 Mt CO<sub>2</sub>/yr annual injection rate, and a fleet of twelve ships (Table 5.4). Hence, taking into account the project lifetime of 25 years, the transport and storage of 20 Mt of CO<sub>2</sub>/per year offshore would require in total from ~€6.3 B to ~€11.9 B.

*Table 5.4: Summary of cost estimations for transporting and storing 20 Mt of CO<sub>2</sub> per year, in a 25-years long offshore project. "Scenario A" represents the cheapest option, whereas "scenario B" the most expensive one.*

|                   | Transport distance | Number of ship | Number of wells | Annual injection rate | TOTAL COSTS |
|-------------------|--------------------|----------------|-----------------|-----------------------|-------------|
|                   | km                 |                |                 | Mt/yr                 | B€          |
| <b>SCENARIO A</b> | 500                | 7              | 7               | 3                     | ~6.3        |
| <b>SCENARIO B</b> | 1500               | 12             | 40              | 0.5                   | ~11.9       |



### 5.5. Costs for CO<sub>2</sub> sequestration in deep-sea basalts

The costs of transport and storage of 20 Mt/yr of CO<sub>2</sub> offshore have been estimated for reservoirs at water depth <2500 m and sediment thickness below the seafloor up to 3000 m [ZEP, 2011a; b]. No calculation has been found concerning reservoirs located in deeper waters, up to 5000 m. Yet, the majority of the oceanic crust areas highlighted for combined gravitational and physical CO<sub>2</sub> trapping are located at water depths greater than 5000 m (Figure 5.2).

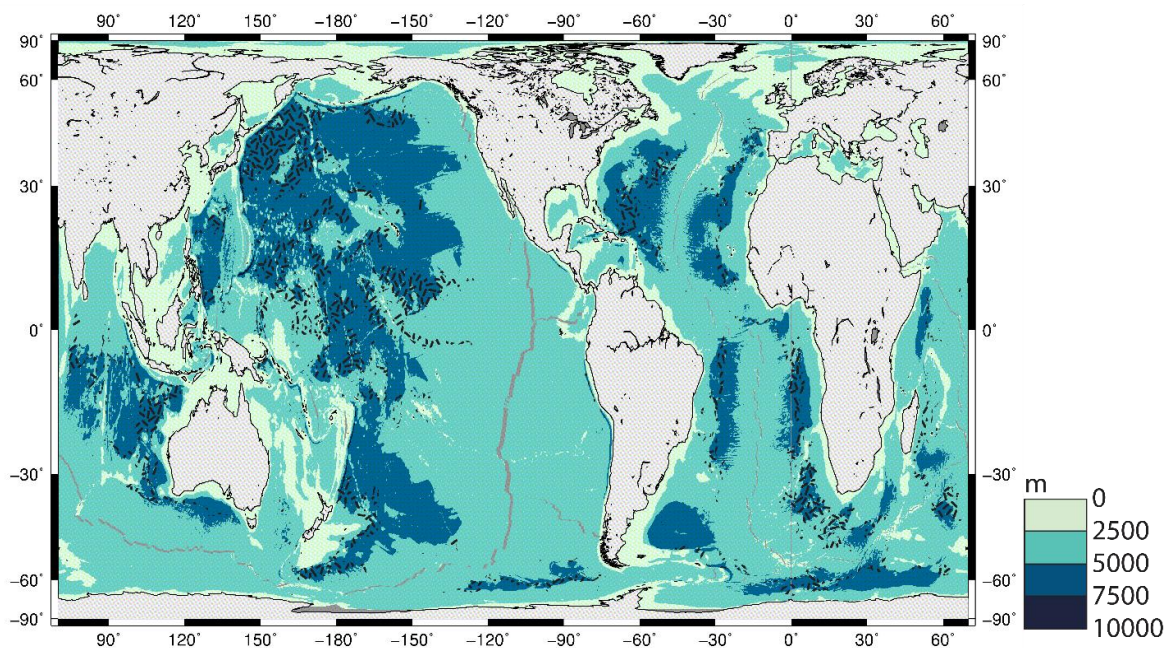


Figure 5.2: Bathymetric map of all the ocean, with contours every 2500 m. Locations for stable geological sequestration of CO<sub>2</sub> at the sediment-basement interface are shown by the black dashes.

However, if only gravitational trapping at the sediment-basement interface (Figure 5.3) is required, the extent of these areas potentially suitable for geological CO<sub>2</sub> storage in deep-sea basalts significantly increases. In particular, a new group of targets become available at shallower water depths, ranging from 2500 to 5000 m, but mostly further offshore. These depths are still greater than those considered in the cost estimations, but more accessible in terms of technological feasibility of CCS in offshore reservoirs.



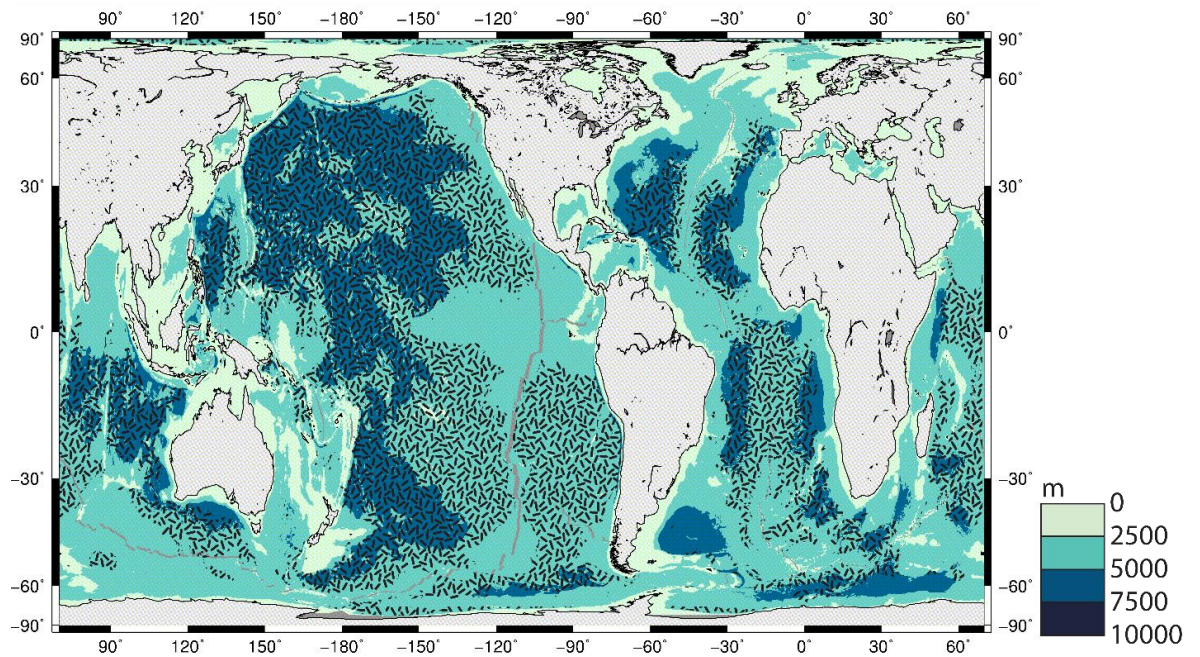


Figure 5.3: Bathymetric map of all the ocean, with contours every 2500 m. Locations for gravitational trapping of CO<sub>2</sub> ( $\rho_{\text{CO}_2} > \rho_{\text{seawater}}$ ) at the sediment-basement interface are shown by the black dashes.

In order to qualitatively scale up the cost estimations to deeper water depths, corresponding to locations for stable geological sequestration of CO<sub>2</sub> in deep-sea basalts, the IODP drilling time estimator [[http://iodp.tamu.edu/participants/coring\\_estimator.html](http://iodp.tamu.edu/participants/coring_estimator.html)] has been used to calculate several scenarios (Figure 5.4). This evaluation is based on the reasonable assumption that the total costs are proportional to the time spent for drilling wells at the storage site. Also, technological development is required to ensure the application of the transport facilities described in this study, particularly the STL system, for greater water depths, up to 5000 m.

For each of three water depths (1000, 2500, and 5000 m), three drilling conditions have been considered: **a**- 200 m of sediments and 300 m of basement, **b**- 3000 m of sediments, **c**- 700 m of sediments and 300 m of basement. Scenario **a** represents the minimum condition of combined gravitational and physical trapping (Chapter 3); scenario **b** the typical conditions of hydrocarbon production fields, which are included in the cost

calculations (Section 5.3); scenario **c** the maximum amount of sediments for combined gravitational and physical trapping (Chapter 3).

To penetrate these rock thicknesses, the IODP calculations consider a rotary drilling (RCB) approach, with penetration rates decreasing from ~30 m/h (for sediments), to ~4.5 m/h (for the first 50 m of basement), ~3.0 m/h (for the next 100 m), and ~2.0 m/h (thereafter).

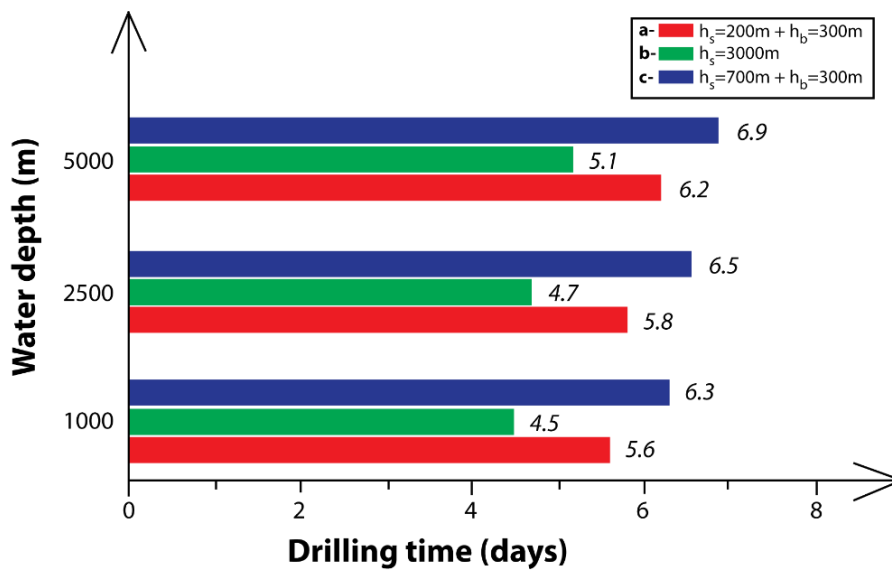


Figure 5.4: Summary of drilling times for several scenarios of water depth, sediment ( $h_s$ ) and basement thickness ( $h_b$ ) [[http://iodp.tamu.edu/participants/coring\\_estimator.html](http://iodp.tamu.edu/participants/coring_estimator.html)].

In general, the IODP drilling time results highlight that the cost associated with drilling of up to 3000 m of sediments (scenario **b**) are lower (~30%) than the costs associated with combined gravitational and physical trapping (scenarios **a** and **c**), for all water depths. On the other hand, the increase in water depth increases the drilling time. Particularly, the transition from 2500 to 5000 m of water leads to an increase in drilling time by a factor of ~1.1 on average (~10%).

Translated into costs, these results suggest that to scale a CCS project up to 5000 m water depth, as in the case of deep-sea basalts, capital costs should increase by ~1.4 times (~40% increase). For example, taking into account a project lifetime of 25 years,

the transport and storage of 20 Mt of CO<sub>2</sub> at a reservoir located 500 km offshore, at ~5000 m of water depth, would require in total a minimum of ~€8.8x10<sup>9</sup> (Table 5.5), which almost corresponds to the cost of the Prelude FLNG (see Section 5.2.). Also, given the total volume of 500 Mt of CO<sub>2</sub>, the cost of storing carbon dioxide for the 25 year-long project in deep-sea basalts at ~5000 m water depth ranges from ~18 €/t (500 km) to ~33 €/t (1500 km). Only the lowest estimation is comparable with the current injection costs in the Sleipner project that correspond to 17 \$ per tonne of CO<sub>2</sub> (~15 €/t) [[www.aapg.org/publications/blogs/energy-policy/article/articleid/12373](http://www.aapg.org/publications/blogs/energy-policy/article/articleid/12373)], and in the Weyburn-Midale project that are 20 \$ per tonne of CO<sub>2</sub> (~18 €/t) [[www.sequestration.mit.edu/tools/projects/weyburn.html](http://www.sequestration.mit.edu/tools/projects/weyburn.html)], with the difference that the latter two projects include the monitoring system. On the other hand, the costs associated with the *ex situ* mineral carbonation option have been estimated to be at least ~54 \$ per tonne of CO<sub>2</sub> (~48 €/t), with separation and transportation costs for CO<sub>2</sub> not included [Gerdemann et al., 2007]. Rubin [2008] determined that the CO<sub>2</sub> capture step accounts for up to ~80% of the whole CCS project costs. For example, the CO<sub>2</sub> captured through the most common post-combustion technique is estimated to cost between 48 and 68 \$ per tonne of CO<sub>2</sub> (~42-60 €/t) [Rubin et al., 2015].

More generally, it is also important to contextualise these cost estimations to better understand the potential CCS role within a global energy market. To date, the price for a barrel of crude oil (~0.136 t of crude oil) is ~\$50 [[www.nasdaq.com/markets/crude-oil](http://www.nasdaq.com/markets/crude-oil)], which corresponds to ~€315 per tonne of oil. Given that 1 tonne of crude oil produces ~3.2 t of CO<sub>2</sub> [EPA, 2015], 1 t of CO<sub>2</sub> is emitted for every ~€100 invested in crude oil. For comparison, the storage of that same tonne of CO<sub>2</sub> would cost between ~12.6 to ~33.3 €/t when associated with a 25 years transport and storage CCS project offshore. Furthermore, the global subsidies given to fossil fuel extracting companies in 2011 amounted to \$4200x10<sup>9</sup> (~€3770x10<sup>9</sup>), and increased to \$5300x10<sup>9</sup> (~€4760x10<sup>9</sup>) in 2015 [Coady et al., 2015]. For purpose of comparison, the total costs needed during the first year of a CCS project, involving the transport and storage of 20 Mt of CO<sub>2</sub>, at

reservoirs located 500 km offshore, and with seven injection wells, at water depth of 2500 m, are  $\sim\text{€}1.7 \times 10^9$ ; the same project has a total cost of  $\sim\text{€}6.3 \times 10^9$  during its whole lifetime of 25 years (Chapter 5). These values corresponds to only  $\sim 0.03$  and  $\sim 0.13\%$ , respectively, of 2015 global subsidies for fossil fuel extraction companies.

*Table 5.5: Comparison of total cost estimations for transporting and storing 500Mt of CO<sub>2</sub> in 25 years in deep-sea basalts located at water depth of  $\sim 2500$  and  $\sim 5000$  m, for two distance scenarios (500 and 1500 km, respectively).*

| Distance<br>from shore | Water depth = 2500 m    |                      | Increasing<br>factor | Water depth = 5000 m    |                      |
|------------------------|-------------------------|----------------------|----------------------|-------------------------|----------------------|
|                        | Total costs             | Cost CO <sub>2</sub> |                      | Total costs             | Cost CO <sub>2</sub> |
| km                     | $\times 10^9 \text{ €}$ | €/t                  |                      | $\times 10^9 \text{ €}$ | €/t                  |
| 500                    | $\sim 6.3$              | $\sim 12.6$          | 40%                  | $\sim 8.8$              | $\sim 17.6$          |
| 1500                   | $\sim 11.9$             | $\sim 23.8$          |                      | $\sim 16.7$             | $\sim 33.3$          |

## 5.6. Conclusions

The majority of the information available to estimate the costs of geological storage of carbon dioxide offshore are restricted to sedimentary reservoirs for hydrocarbon production, due to the difficulties and high costs associated with offshore EOR projects. In this chapter, a general estimation of the costs for transport and storage of CO<sub>2</sub> offshore has been attempted, together with a number of limitations and constraints on its applicability on a broader scale. The results can be summarised as follow:

- the CO<sub>2</sub> costs linked with ship transport and storage in offshore formations can be estimated for reservoirs at water depth < 2500 m and assuming injection wells are drilled into sediments;
- the suitable sites for combined gravitational and physical trapping at the sediment-basement interface are located in water depths >5000 m, whereas some of the areas defined by gravitational trapping are characterised by bathymetry between 2500 and 5000 m;
- the total costs for a 25 year-long CCS project, involving the transport and storage of 20x10<sup>6</sup> t of CO<sub>2</sub> per year at reservoirs located 500 to 1500 km offshore, in water depth up to 2500 m, are estimated to range from ~€6.3x10<sup>9</sup> to ~€11.9x10<sup>9</sup>;
- assuming proportionality between cost and drilling time, the corresponding total costs for a 25 year-long CCS project scaled up to 5000 m of water depth have to be multiplied by a factor of ~1.4;
- the transport and storage of 500 Mt of CO<sub>2</sub> over 25 years in a deep-sea basaltic reservoir located 500 km offshore, at water depths of 5000 m, would require in total a minimum of ~€8.8x10<sup>9</sup> for capital and operational costs, and the cost of carbon dioxide for the whole project would be ~18 €/t.



## Chapter 6: Discussion and conclusions

### 6.1. Results summary

The primary goal of this thesis was to better understand the potential of the oceanic crust as a reservoir to store anthropogenic CO<sub>2</sub>. For this purpose, several methodologies were applied: A) to illustrate the phase relationship of carbon dioxide in the seafloor; B) to explore the feasibility of carbon sequestration in deep-sea basalts based on gravitational and physical trapping at the sediment-basement interface; C) to determine the dissolution extent and rate of dissolution of oceanic rocks in batch experiments at low T, pCO<sub>2</sub> of 1 atm, and using seawater as a solution; and D) to explore the economic costs of transport and storage of CO<sub>2</sub> in offshore reservoirs.

The main outcomes of this work are: 1) carbon dioxide is denser than seawater only for pressures greater than 27 MPa (~2700 m of water) and temperatures between 0 and 30 °C; 2) regions characterised by thermodynamically stable liquid CO<sub>2</sub> are generally located in old oceanic crust covered by a relatively thin layer of sediments; 3) the combination of gravitational and physical trapping limits the oceanic crust suitable for CCS to ~8% of its area, but highlights five potential targets where even the smallest identified reservoir could provide sufficient carbon dioxide sequestration capacity for several centuries of anthropogenic CO<sub>2</sub> output; 4) based on the low temperature batch experiments in a CO<sub>2</sub>-seawater-rock system, Ca and Si are key elements detecting the dissolution of cations from the solid to the aqueous solution; 5) plagioclase and, if present, amphibole and calcite are the reactive minerals providing Ca in the batch dissolution experiments, and their composition determines the different Ca release rates of studied rocks; 6) the total costs for CCS projects with a lifetime of 25 years, involving

the transport and storage of 20 Mt of CO<sub>2</sub> per year, in deep-sea basaltic reservoirs located 500 and 1500 km offshore, respectively, and in water up to 5000 m deep, are estimated to be ~€8.8x10<sup>9</sup>, and ~€16.7x10<sup>9</sup>, respectively; 7) the corresponding total costs per tonne of CO<sub>2</sub> transported and stored are estimated to be ~18 €/t, and ~33 €/t, respectively.

## **6.2. CO<sub>2</sub> trapping mechanism implications**

The three trapping mechanisms proposed to geologically store the CO<sub>2</sub> in offshore mafic and ultramafic formations are: gravitational (Chapter 2), physical (Chapter 3), and geochemical. In this study, only the dissolution reactions involved with the early stages of geochemical trapping have been investigated (Chapter 4).

In this thesis, the maximum storage stability is reached with the combination of gravitational and physical trapping mechanisms (Chapter 3), but at these conditions the CO<sub>2</sub> sequestration in deep-sea basalt does not represent an economically viable option (Chapter 5, and Section 6.5). However, the advantages and disadvantages of these mechanisms are discussed individually.

If only a thick layer of sediments above the oceanic basement, ranging from 200 to 700 m (*physical trapping*), is considered as a trapping mechanism, ~32% of the seafloor is suitable for CO<sub>2</sub> storage in mafic formations (Chapter 3). However, much of this region is characterised by temperatures and pressures that make CO<sub>2</sub> much less dense than seawater at the sediment-basement interface (Chapter 2), due to the shallower bathymetry. This means that the CO<sub>2</sub> would have a tendency to rise towards the seafloor, driven by buoyancy, instead of sinking into the basement. Also, the locations showing effective physical trapping would have to be validated in terms of cap rock integrity, permeability, stratigraphy, and hydrodynamic fluid dispersion (Chapter 1).



On the other hand, if the only requirement for CO<sub>2</sub> storage is that CO<sub>2</sub> is denser than seawater at the sediment-basement interface (*gravitational trapping*), as much as ~48% of oceanic crust is available (Chapter 3). CO<sub>2</sub> injection at these sites would have the advantage of trapping the CO<sub>2</sub> in the basement. These circumstances, together with the presence of water depth up to ~5000 m for reservoirs ~500 km from the coast, or shallower water for reservoirs up to 1500 km from the shore, could represent more economically feasible scenarios for the application of CCS offshore (Chapter 5). For example, a more detailed investigation of *target a* (offshore Western Australia) highlights the presence of several square kilometres of oceanic crust, located <500 km from the southern coast of Western Australia in deep waters up to ~5000 m (*target a1* - Chapter 3), that satisfies the more economical conditions for gravitational trapping only (Chapter 5).

In this thesis, the application of combined physical and gravitational trapping in offshore reservoirs as a CO<sub>2</sub> storing mechanism, independently from the mineral carbonation potential of basalts, identifies numerous suitable locations (Chapter 3), with CO<sub>2</sub> stable in a liquid state. However, these locations would not provide ideal conditions for geochemically trapping the CO<sub>2</sub>. The high pressures, low temperatures, and liquid phase of carbon dioxide described in this thesis (Chapter 2) would create solubility issues, based on literature data. In a liquid state, CO<sub>2</sub> takes over hundreds of years to dissolve in the formation brine [Hirai *et al.*, 1997; House *et al.*, 2006]. A potential way to overcome this problem is to accelerate the dissolution through mixing the CO<sub>2</sub> in seawater prior to its injection – solubility trapping (Chapter 3) [Gislason and Oelkers, 2014; Matter *et al.*, 2011]. This process was partially simulated in the batch experiments, where gaseous CO<sub>2</sub> was maintained fully dissolved in solution (Chapter 4). However, the suggested approach involves i) very large quantities of seawater, ii) the presence of porous rocks to not over pressurise the reservoir with the huge amount of pumped water, and iii)

increase in storage costs due to the process of dissolving CO<sub>2</sub> in solution prior to its injection (Chapter 3).

### **6.3. Carbon sequestration on mid-ocean ridge flanks**

One of the main reasons to look at the oceanic crust as a carbon storage reservoir is linked to the vast size of porous basement, where CO<sub>2</sub> can be fixed in mafic and ultramafic rocks through carbonate mineral formation (Chapters 2 and 4). Many studies on natural analogues, represented by carbonate veins within the basement, suggest the mid-ocean ridge flanks are potential locations for CO<sub>2</sub> sequestration offshore, because of hydrothermal circulation, and the sufficient porosity and permeability within the extrusive section (Chapters 1 and 2). Old oceanic crust in deep oceans shows a great potential for combined gravitational and physical CO<sub>2</sub> trapping (Chapters 2 and 3). On the other hand, most of the areas close to the ridge axis – which is characterised by hot, freshly created oceanic crust – or nearby continental margins – which generally include thick layers of sediments above the basement – are considered unsuitable for combined CO<sub>2</sub> trapping because of the negative  $\Delta p$  conditions (Chapters 2 and 3).

A detailed investigation on the eastern flank of the Juan de Fuca (JdF) ridge, was done because it was proposed as a potential offshore target for carbon dioxide storage [Goldberg *et al.*, 2008; Slagle and Goldberg, 2011]. Previous studies have shown that carbonate veins formed from low temperature basement fluids (<70 °C) present in porous rocks [Coggon *et al.*, 2004], and therefore that carbon sequestration reactions occur naturally *in situ*. In this project, batch dissolution experiments carried out at 40 °C on JdF samples detected Ca and Si dissolution rates at pH ~5 (from -14.1 to -14.9 log (mol/cm<sup>2</sup>/s), and from -14.7 to -14.9 log (mol/cm<sup>2</sup>/s), respectively), which are comparable with highly reactive crystalline basalt from Iceland (from -15.0 to -15.5 log (mol/cm<sup>2</sup>/s), and from -14.5 to -15.2 log (mol/cm<sup>2</sup>/s), respectively) [Gudbrandsson *et al.*, 2011]

(Chapter 4). Another advantage put forward for JdF as potential offshore target is the presence of relatively continuous and anomalously thick sediment cover, particularly efficient for physical trapping (Chapter 2) [Goldberg *et al.*, 2008]. However, this study shows that the JdF site is not suited for CCS purposes, based on gravitational trapping, because the local conditions are too hot to keep the CO<sub>2</sub> within the basement ( $\rho_{\text{CO}_2} > \rho_{\text{seawater}}$ , Chapter 3).

On this basis, the mid-ocean ridge flanks proximal (~500 km) to the ridge axes are rarely included in the maps proposed for the combined gravitational and physical trapping (Figure 3.7, Chapter 3). On the other hand, these same mid-ocean ridge flanks become suitable locations for CO<sub>2</sub> sequestration when only gravitational trapping is applied (Figure 3.3, Chapter 3). Despite sediment thicknesses <200 m, the high water depth at open ocean conditions (from 2500 and 7500 m) is sufficient to keep the pressure high enough for the CO<sub>2</sub> to be gravitationally stable. Nevertheless, these gravitationally suitable locations do not correspond to economic viable options due to the increased cost mainly associated with the distance from the coast (Chapter 5).

#### **6.4. Batch CO<sub>2</sub> dissolution experiment implications**

The experimental work conducted for this thesis provides important insight into the determination of dissolution reactions of basalts and gabbro from the modern ocean floor and ophiolites, respectively, in the presence of CO<sub>2</sub>-rich seawater solution (Chapter 4). Batch reactor experiments (Chapter 4) revealed that the rate of dissolution of crystalline mafic rocks in a CO<sub>2</sub>-rich seawater solution can be detected primarily through silicon concentrations, which is consistent with previous experiments, and also through calcium concentrations. This finding is a step forward in the understanding of mineral reactivity within the rock. In fact, the highest whole rock Ca dissolution rates were measured from the JdF samples with ~1% of calcite (mix 1 and mix 2), but also from the ophiolitic gabbro

(G1), for which the only Ca potential source is amphibole and plagioclase, as almost no pyroxene is present. This latter observation leads to the hypothesis that differences in calcium release rates are related to calcite, amphibole and plagioclase content (Chapter 4). Also, anorthitic plagioclase (Ca ~1 mol) reacts faster than labradorite (Ca ~0.6 mol) and oligoclase (Ca ~0.2 mol), as shown in previous studies [e.g., Casey *et al.*, 1991; Gudbrandsson *et al.*, 2014; Oxburgh *et al.*, 1994]. Hence, plagioclase composition is a potential key factor defining the Ca-release rate (i.e. dissolution rates) during the low temperature (40 °C) experiments.

In a more general scenario, bearing in mind that the availability of cations (such as Ca, Mg, and Fe) in solution is considered one of the major challenges in the mineral sequestration of CO<sub>2</sub> [Oelkers *et al.*, 2008], this observation may lead to one of the most straightforward methods to distinguish which mafic formation should be taken into account for mineral carbonation for CCS purposes, offshore as well as on land. For example, the old oceanic crust (155 Ma) in the Indian Ocean – offshore Western Australia (*target a*, Chapter 3), already identified to be suitable for combined physical and gravitational trapping – also shows a theoretically high geochemical potential. In fact, the basement of *target a* presents favourable characteristics highlighted above: cation sources provided by phenocrysts of olivine (fast reaction rates [Kelemen and Matter, 2008]) and plagioclase as calcic as An<sub>90</sub> (90% anorthitic plagioclase) [Ishiwatari, 1992]; a nonpervasive low temperature alteration, typical of seafloor weathering, which confirms the past occurrence of carbonation reactions [Gillis *et al.*, 1992]; and a porosity on average around 5% [Brereton, 1992] to host the CO<sub>2</sub>-rich fluids.

## **6.5. CO<sub>2</sub> cost implications for CCS offshore**

Cost analysis of a potential CCS project with a reservoir located offshore highlighted two main factors that influence the estimations: 1) water depth, and 2) distance offshore.

Water depth is a major issue for the technologies considered in this study. In fact, ships with unloading buoys have been designed for bathymetric values up to 2500 m, but the proposed targets for CO<sub>2</sub> storage in basalts offshore are usually much greater than 2500 m for both cases of combined trapping and gravitational-only trapping (Chapters 3 and 5). Assuming improvements in future technology, and costs proportional to drilling time, the estimated costs for a CCS project in 5000 m of water depth were scaled up by a factor of 1.4 (Chapter 5). Consequently, a 25 year project involving the transport and storage of 20x10<sup>6</sup> t of CO<sub>2</sub> per year to a reservoir located 500 km from the shore, at ~5000 m of water depth, with seven drilled wells, would require in total ~€8.8x10<sup>9</sup> (Figure 6.1). This estimation is ~€5x10<sup>9</sup> lower than the cost calculated for the corresponding option with 40 drilled wells (~€13.4x10<sup>9</sup>), and ~3x10<sup>9</sup> lower than the option at 2500 m of water depth, 1500 km offshore, and 40 drilled wells (~€11.9x10<sup>9</sup>) (Figure 6.1).

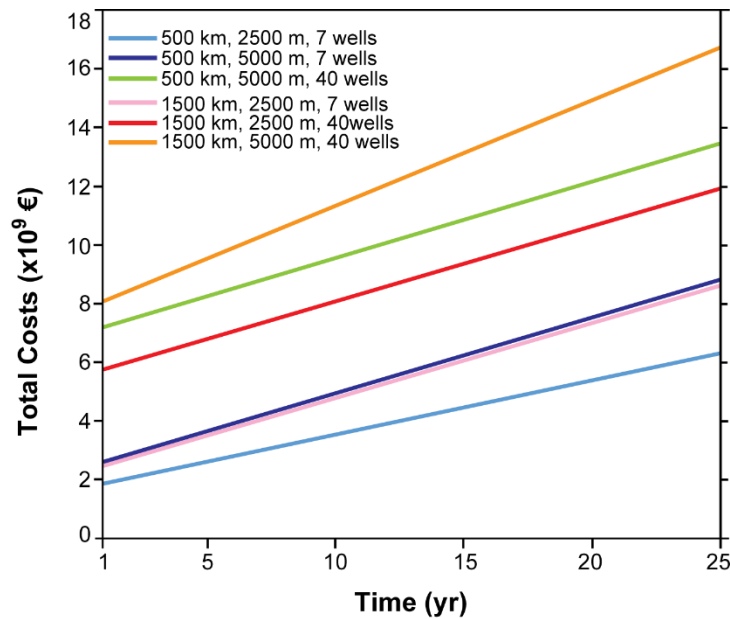


Figure 6.1: Summary of cost estimations for the transport and storage of 20 Mt of CO<sub>2</sub> in 25 years-long CCS projects. Six scenarios are shown as a function of distance from shore (500 or 1500 km), water depth (2500 or 5000 m), and number of injection wells (7 or 40).

Hence, these evaluations show that a CCS project is more economically viable when closer to the shore, even if in deeper waters, and using fewer wells, than if located further away, in shallower waters, and using many wells. The number of wells depends on the

injection rate, which in turn is linked to the reservoir pressure and its permeability (Chapter 5). A high injectivity is required to reduce the number of wells, hence making deep water projects more attractive.

The influence of the distance from the coastline on the costs is intuitive: a greater distance to cover during the CO<sub>2</sub> transport corresponds to a higher number of ships required in the CCS fleet (Chapter 5), and therefore requiring larger volumes of fuel to operate them. Also, it is important to bear in mind that every step in the CCS chain also generates CO<sub>2</sub>: for example, the transport of CO<sub>2</sub> via ship. The design of CO<sub>2</sub> carriers considered during the cost analyses (40,000 m<sup>3</sup>, -50 °C, and ~7 bar – Chapter 5) is very similar to that of low temperature Liquid Petroleum Gas (LPG) carriers, which on average emit 43.5 g CO<sub>2</sub>/tonne-km [*Second IMO GHG Study 2009*]. A one way journey of a ship involved in a CCS project (ship capacity of 0.046 Mt of CO<sub>2</sub> per journey) with targeted reservoir at 500 km offshore (~22 h journey) produces ~1000 t of CO<sub>2</sub>.

Hence, the definition of the most economic location for a reservoir offshore is not straightforward, and further assessments are required to determine the capacity of specific reservoirs and their size. For example, assuming a basement thickness of 300 m and relative rock porosity of 10%, estimated for combined physical and gravitational trapping (Chapter 3), only 16 km<sup>2</sup> of mafic rocks would be required to store 500 Mt of CO<sub>2</sub> (at average density of 1066 kg/m<sup>3</sup>) in 25 years.

## **6.6. Social implications**

Anthropogenic CO<sub>2</sub> emissions into the atmosphere must be drastically reduced to limit the increase of global average temperature to less than 1.5 °C above pre-industrial levels [*Conference of the Parties Twenty-first Session, 2015*] (Chapter 1). This will require a combination of several strategies [*IPCC, 2005b; 2014a*]. Improved energy efficiency, and reduction in the use of fossil fuels as a source of energy are probably two of the wiser

options, but the acceptance of the latter case by the general public, which is key for successful implementation, is a constant issue [Judkins *et al.*, 1993]. Increasing the carbon sink is an alternative strategy, and CCS is one of the most viable options [IEA, 2010; Wright *et al.*, 2014] (Chapters 1). However, little is known about the public perception of CCS as a mitigation strategy. Data on the topic have only been explored at small scale, and are technique-specific [Shackley *et al.*, 2009]. Also, most of the surveys are carried out for ongoing projects that are affiliated to Enhanced Oil Recovery – EOR, and deep-sedimentary aquifers [Scott, 2013]. These studies all found quite low levels of public awareness of CCS, and more generally of the perceived urgency of addressing climate change. Onshore CO<sub>2</sub> storage is suggested to cause the greatest concern, whereas, for the offshore projects, the main fears are associated with the CO<sub>2</sub> pipelines in the ocean. Overall, it appears that CCS would be more accepted when combined with other low-carbon strategies.

This thesis presents an unconventional approach to CCS independent from the petroleum industry, and by taking into account the stability of carbon dioxide in oceanic crust (Chapter 3). Also, an attempt to constrain the costs associated with CO<sub>2</sub> transport via ship, and its storage offshore, has been carried out to justify CCS strategies to a broader audience, comparing the main findings with oil industry costs (Chapter 5). Although the CO<sub>2</sub> costs for a whole transport-storage CCS project of 25 years is lower than current oil costs (Section 6.5), CCS strategies are still not playing a major role in the global market. The carbon tax on CO<sub>2</sub> emissions, so far implemented by only fifteen countries [[www.worldbank.org/content/dam/Worldbank/document/Climate/background\\_notecarbon-tax.pdf](http://www.worldbank.org/content/dam/Worldbank/document/Climate/background_notecarbon-tax.pdf)], ranges from 5 to 150 € per tonne of CO<sub>2</sub> equivalent (measure used to compare the emissions from various greenhouse gases based on their warming potential relative to that of carbon dioxide), and has been one of the main drivers for starting commercial CCS projects, besides EOR, such as the Sleipner project. Hence, is

this carbon tax going to be applied worldwide? If so, will it be sufficient to boost the CCS market? Will its implementation change the public perspective on climate change?

The answers go beyond the scope of this thesis, but it is important to keep in mind the social implications of a research that focuses on such a complex challenge like the CCS.

## 6.7. Limitations

As this thesis research is focused on CCS offshore, a number of limitations are associated with the availability of relevant data (e.g. measurements from scientific drilling cruises). A series of reasonable assumptions was put in place to overcome this problem on a global scale, but further investigations are required. The database-related limitations are listed below, together with potential solutions.

- 1) *Temperature validation*: despite numerous holes drilled into the oceanic crust during scientific oceanic cruises, validation of the estimated temperature at the sediment-basement interface has only been done at thirteen sites (Chapter 2).
- 2) *Sediment thermal conductivity*: an average thermal conductivity value ( $K_s = 1 \text{ W/(m K)}$ ) was taken into account for all the sediment layers covering the whole oceanic crust, justified by global databases of marine sediment thermal conductivities (Chapter 2). An exception was made for the Juan de Fuca Plate, where more detailed investigations revealed a locally more appropriate thermal conductivity of  $2 \text{ W/(m K)}$  (Chapter 2).
- 3) *Basement permeability*: the volume of storable  $\text{CO}_2$  in each of the identified targets was estimated assuming values of porosity (10%) and thickness (300 m) of basement, but without considering the permeability (Chapter 3), which within the uppermost few hundred meters of the igneous crust ranges from  $10^{-22}$  to  $10^{-9} \text{ m}^2$  [Fisher, 1998]. Its local variability and lack of measurements recorded during scientific cruises make its inclusion in the global investigations difficult.



In general, all the above limitations could be solved in the future with additional measurements from scientific cruises, more localised data analyses, and potential field-scale projects in the areas of interest.

Other important limitations of this study are associated with the experimental nature of the geochemical investigation (Chapter 4).

A) *Geochemical process*: the experiments were focused on dissolution reactions only to constrain the dissolution potential of oceanic rocks exposed at low temperature to a CO<sub>2</sub>-rich seawater. Hence, in this far-from equilibrium setup, no further reactions, such as precipitation, were possible.

B) *Fine particles*: the decision to use oceanic cores involved the disadvantage of having less material available to start with, which in turn led to a compromise during the cleaning procedure, and the subsequent presence of fine particles in these samples. However, physical analyses were carried out to constrain the sample particle size, to account for their effect on the reaction kinetics, and to allow comparison with other experiments.

C) *Experimental settings*:

- the chosen temperature of 40 °C, and the continuous flux of CO<sub>2</sub> in and out of the reactor to maintain the saturation conditions led to the partial evaporation of the seawater. This was corrected by the consideration of the chloride concentration.
- the observation of corrosion on the stainless steel bubbler meant that analyses of iron and aluminium were compromised. Note that the only alternative gas distribution tube available in the market is made of borosilicate glass, which could compromise the measurement of silicon in the system;
- the reaction extent, most probably associated only with the top-surface of the samples, could be improved by stirring the solution, ensuring that no further decrease in the sample particle size will take place.

## 6.8. Future work

This research covers a number of scientific questions, ranging from the use of geophysical models for oceanic crust, to experimental work on geochemical reactions, and to economic evaluations on CCS feasibility offshore. This interdisciplinary approach has been chosen above a more specific one because of the CCS nature itself, which is complex and requires a broader assessment of risks, particularly for large-scale CCS decisions [Choptiany *et al.*, 2014]. Consequently, further work on each of these topics could be developed in the future.

Investigations on physical and gravitational CO<sub>2</sub> trapping at the sediment-basement interface for the whole oceanic crust show the sequestration potential of several areas on a global scale. However, more detailed analyses on Juan de Fuca Plate and eastern equatorial Pacific Ocean highlight the need of validating the general oceanic crust properties to a more localised level of research in order to define the real potential of each reservoir. As for the case of *target a* (NW Western Australia), all the other identified targets could undergo specific examinations. Also, more parameters could be included in the evaluation if, in any of the areas of interest, *in situ* measurements of permeability and sediment thermal conductivity become available.

The preliminary results from the experiments on several oceanic rocks highlighted the possibility of improving the experimental design, without major changes. For example, further experiments could be carried out as follows:

- a) with 5-10 g of material, to have the sample always in contact with the CO<sub>2</sub>-rich solution, and to better define the reactive mass;
- b) with only anorthitic plagioclase, or actinolitic amphibole, to validate the hypothesis on the dependency of the mafic rock reactivity from these mineral compositions.

However, for a complete understanding of the rock reactivity, these samples should be exposed to different solutions, and type of experiments (such as flow-through, and combined dissolution-precipitation), but this latter option would require major efforts.

In general, a geochemical model based on the CO<sub>2</sub>-seawater-oceanic rocks used during the dissolution experiments could provide more insight into the efficacy of mineral dissolution reactions, with particular attention to the behaviour of secondary minerals. Regarding CO<sub>2</sub> costs, the only way to improve the theoretical estimations would be through the possibility to access information of already ongoing offshore CO<sub>2</sub> storage aspects, including those related to EOR, or being involved in an offshore field-scale injection project for CO<sub>2</sub> storage in mafic or ultramafic rocks.

## 6.9. Conclusions

At a time when environments are fast changing with global warming extremely likely due to the observed increase in anthropogenic greenhouse gas concentrations [IPCC, 2014b], mitigation strategies are the focus of increasing attention. The approaches proposed to counteract the levels of carbon dioxide in the atmosphere can be divided in “reducing emissions” methodologies (such as reducing energy demand, improving technology efficiency, and increasing the contribution of nuclear and low carbon renewable energy), and in “increasing carbon sinks” practices (such as Carbon Capture and Storage – CCS).

Understanding the advantages and limits of carbon capture and storage, from both economic and scientific perspectives, is essential for governments, stakeholders, academics, and the general public. In the last decades, CO<sub>2</sub> injections into mafic and ultramafic formations have started to be considered as an alternative to more conventional storage in sedimentary basins (e.g. CarbFix project, and Big Sky Carbon Sequestration Partnership). However, the storage capacity of these reservoirs has yet to be determined. The present thesis focused on the sequestration process of the CCS chain in a specific context, which involves the geological storage of carbon dioxide in deep-sea basalts. The thermodynamic properties at the sediment-basement interface for the whole oceanic crust, together with rock dissolution rates in a CO<sub>2</sub>-seawater-oceanic rock system, and costs related to CO<sub>2</sub> transport offshore and relative underground storage were investigated. The outcomes constrained the viability of geological sequestration of CO<sub>2</sub> in oceanic crust via combined physical and gravitational trapping, proposing several suitable reservoirs, where even the smallest target could contribute to the reduction of several centuries of current anthropogenic CO<sub>2</sub> emissions (~36 Gt of CO<sub>2</sub> per year). On the other hand, this thesis also highlighted how much further work is required to turn this huge theoretical potential into a financially attractive option.

# Appendix

## A-Chapter 1

CH1-1: Marieni et al. (2013).

### Geological storage of CO<sub>2</sub> within the oceanic crust by gravitational trapping

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Received 27 October 2013; accepted 4 November 2013.

[1] The rise of atmospheric carbon dioxide (CO<sub>2</sub>) principally due to the burning of fossil fuels is a key driver of anthropogenic climate change. Mitigation strategies include improved efficiency, using renewable energy, and capture and long-term sequestration of CO<sub>2</sub>. Most sequestration research considers CO<sub>2</sub> injection into deep saline aquifers or depleted hydrocarbon reservoirs. Unconventional suggestions include CO<sub>2</sub> storage in the porous volcanic lavas of uppermost oceanic crust. Here we test the feasibility of injecting CO<sub>2</sub> into deep-sea basalts and identify sites where CO<sub>2</sub> should be both physically and gravitationally trapped. We use global databases to estimate pressure and temperature, hence density of CO<sub>2</sub> and seawater at the sediment-basement interface. At previously suggested sites on the Juan de Fuca Plate and in the eastern equatorial Pacific Ocean, CO<sub>2</sub> is gravitationally unstable. However, we identify five sediment-covered regions where CO<sub>2</sub> is denser than seawater, each sufficient for several centuries of anthropogenic CO<sub>2</sub> emissions. **Citation:** Marieni, C., T. J. Henstock, and D. A. H. Teagle (2013), Geological storage of CO<sub>2</sub> within the oceanic crust by gravitational trapping, *Geophys. Res. Lett.*, 40, doi:10.1002/2013GL058220.

#### 1. Introduction

[2] Human activities since the industrial revolution have increased atmospheric concentrations of greenhouse gases, in particular carbon dioxide (CO<sub>2</sub>), requiring the development of mitigation strategies to minimize the effect on the global climate and potential ocean acidification [Intergovernmental Panel on Climate Change, 2007]. Various strategies have been proposed to reduce CO<sub>2</sub> emission including reducing energy demand, increasing renewable energy, and carbon capture and storage (CCS) underground. The effectiveness of geological reservoirs depends on their storage capacity, reservoir stability, risk of leakage, and the retention time [Hawkins, 2004; Rochelle et al., 2004], with deep saline sedimentary aquifers [Eccles and Pratson, 2012; House et al., 2006; Levine et al., 2007; Schrag, 2009], and depleted oil and gas reservoirs

[Bachu, 2000; Jessen et al., 2005] receiving the greatest research attention. In addition, several mafic and ultramafic formations are under consideration for CO<sub>2</sub> storage including lava flows on Iceland [Gislason et al., 2010; Oelkers et al., 2008] and the Columbia River Basalts in the United States [McGrail et al., 2006].

[3] This paper investigates the geological storage of CO<sub>2</sub> in the deep-sea basalts [Goldberg et al., 2008, 2010; Matter et al., 2007; Slagle and Goldberg, 2011] that form the uppermost igneous lavas of the oceanic crust and cover approximately 60% of Earth's surface. These formations may have advantages over other potential geological storage options: (a) large reservoir capacities; (b) low risk of postinjection leakage due to low permeability sediment blankets in some regions; (c) in situ availability of water; and (d) estimated fluid retention times greater than 500 years [Goldberg et al., 2008]. The Juan de Fuca Plate (JdFP), offshore Washington State, has been the focus of conceptual studies of deep-sea basalt CCS [Goldberg et al., 2008] because it is the best studied mid-ocean ridge flank with well-characterized regional thermal and hydrological regimes [Fisher and Davis, 2000]. The rocks on this plate are relatively young, having formed at the Juan de Fuca Ridge less than 11 Myr ago. The pillow lavas that form the upper few hundred meters of the JdFP crust have high connected porosity (>10%) [Fisher, 1998], and the ridge flank is blanketed by a thick (from 30 to over 700 m) sequence of hemipelagic and turbiditic sediments derived from the North American continent.

[4] Other well-studied regions of the upper oceanic crust that have been considered for CO<sub>2</sub> storage [Slagle and Goldberg, 2011] are in the eastern equatorial Pacific Ocean (eePO) at Sites 504 and 1256 located in ~7 and 15 Myr old crust, respectively. Numerous studies on hydrothermal circulation provide detailed descriptions of physical properties and the porosity of the extrusive section of the ocean crust at these sites [Alt et al., 1993; Teagle et al., 2006; Wilson et al., 2006].

[5] Three primary trapping mechanisms for the long-term storage of carbon dioxide in seafloor basalts have been proposed: (1) *Gravitational trapping* under pressure and temperature conditions where CO<sub>2</sub> is more dense than seawater [Levine et al., 2007]. (2) *Physical or permeability trapping*, where the presence of ≥200 m of overlying low permeability marine sediments isolate the CO<sub>2</sub> injected into the basalts from the oceans, so that any leakage is trapped in the sediments [Goldberg and Slagle, 2009]. (3) *Geochemical trapping*, where the CO<sub>2</sub> and water react with the basalt host rocks to form geologically stable carbonate minerals [Matter et al., 2007].

[6] In this study we consider the global variability of sediment thickness, pressure (*p*), and temperature (*T*), and consequently, the density (*ρ*) of CO<sub>2</sub> and seawater at the sediment-basement interface of the oceanic crust, to identify potential targets for combined gravitational and physical CO<sub>2</sub>

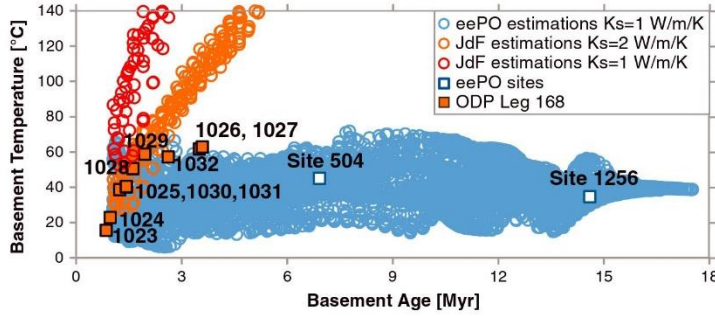
Additional supporting information may be found in the online version of this article.

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**Figure 1.** Comparison between estimated temperatures in the eastern equatorial Pacific Ocean (eePO) and the Juan de Fuca Plate (JdFP), and measured downhole temperatures at the sediment-basement interface. White squares: data from eePO [Alt et al., 1993; Teagle et al., 2006]; orange squares: data from JdFP [Davis et al., 1997]. Circles: estimated values in the eePO (blue), and on the JdFP (red with  $K_s = 1$  W/m/K; orange with  $K_s = 2$  W/m/K).

sequestration. Although this is a physically robust scoping study, detailed programs of local data acquisition are imperative before any targets could be further developed.

## 2. Physical Parameters

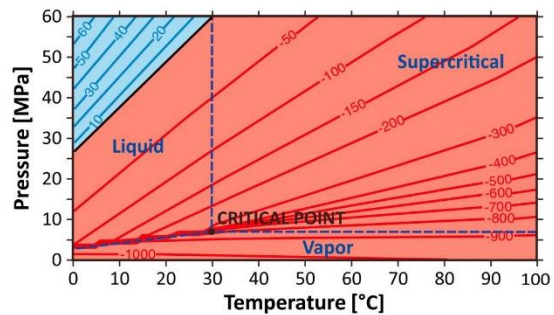
[7] We have developed global maps of the density of seawater and CO<sub>2</sub> at the sediment-ocean crust interface. Pressure was estimated using the NOAA-gridfive sediment thickness database [Divins, 2003] combined with the General Bathymetric Chart of the Oceans (GEBCO)-gridfive world bathymetry map (both 6' × 6' grids) [IOC et al., 2003]. The anomalies in the topography (e.g., abyssal hills and seamounts) are not always detected by global altimetry and gravity analyses, but the NOAA database is the currently best available. We assume a hydrostatic load to the top of basement, using a constant seawater density of 1030 kg/m<sup>3</sup> and a constant salinity of 35 psu (practical salinity unit) [Brown et al., 1995]; this gives a lower bound on the pressure and the CO<sub>2</sub> density. We used the Global Depth and Heat flow model (GDH1) [Stein and Stein, 1992] to estimate the heat flow, and then the temperature, based on the oceanic crustal age [Müller et al., 2008] (see supporting information “Text S1” for the equations). We use GDH1 because it is better for predicting the heat flow at old oceanic crustal ages than the Half Space Cooling Model (HSCM) [Turcotte and Schubert, 2002], and because the Global Heat Flow Database [Pollack et al., 1993] is sparse and irregular. Nevertheless, our GDH1-based approach still presents uncertainties due to incomplete information on sediment thicknesses, local heat flow anomalies, and the thermal properties of sediments.

[8] From the heat flow we have calculated the geothermal gradient [Heberling et al., 2010] and consequently estimated the temperature at the top of the basement (in °C), with the thermal conductivity of the sediments taken as  $K_s = 1$  W/m/K [Pollack et al., 1993; Pribnow et al., 2000a] (see supporting information “Figure S2”).

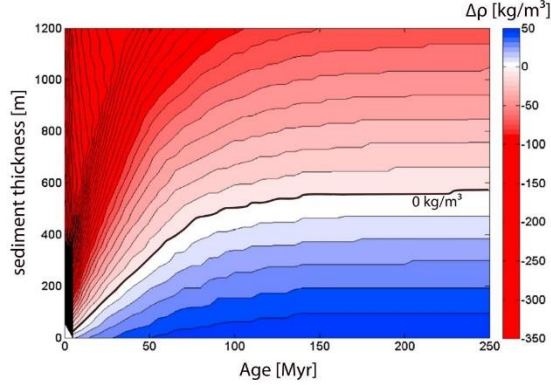
[9] We have validated our estimates of temperature at the sediment-basement interface by using borehole temperature logs from Sites 504 and 1256 [Alt et al., 1993; Teagle et al., 2006] and the Juan de Fuca Plate [Davis et al., 1997] (Figure 1; supporting information “Figure S1” for comparison with HSCM). The two areas have different trends of basement-sediment interface temperature as a function of

age. In the eastern equatorial Pacific Ocean there is good agreement between estimated and measured temperature. However, on the Juan de Fuca Plate, the temperatures at most sites are better fit using the higher measured thermal conductivity ( $K_s \sim 2$  W/m/K) of the local muddy and sandy turbiditic sediments [Pribnow et al., 2000b]. At Sites 1026 and 1027, the measured temperature is lower than predicted because of hydrothermal circulation [Hutnak et al., 2006; Wheat et al., 2004] linked to surrounding basement outcrops (e.g., Baby Bare) [Fisher et al., 2003]. Although there are numerous holes drilled into the oceanic crust by scientific ocean drilling (Deep Sea Drilling Project (DSDP), Ocean Drilling Program (ODP), and Integrated Ocean Drilling Program (IODP)), eePO, and JdF are the only locations where the temperatures close to the sediment-basement interface are well constrained. The validation at the eePO and JdFP gives us confidence in our calculations but emphasizes the need for verification of local physical properties.

[10] The densities of CO<sub>2</sub> ( $\rho_{CO_2}$ ) and seawater ( $\rho_{seawater}$ ) were calculated for pressures from 0 to 60 MPa and temperatures from 0 to 100°C. The CO<sub>2</sub> density was determined by interpolating the online National Institute of Standards and



**Figure 2.** Density difference ( $\Delta\rho = \rho_{CO_2} - \rho_{seawater}$ , in kg/m<sup>3</sup>) between CO<sub>2</sub> and seawater density as a function of pressure between 0 and 60 MPa, and temperature between 0 and 100°C, with the phase diagram of CO<sub>2</sub> overlaid. Positive differences shown in blue indicate conditions for gravitational trapping.



**Figure 3.** Density difference between CO<sub>2</sub> and seawater at the sediment-basement interface as a function of plate age and sediment thickness using the GDH1 model to determine both water depth and thermal conditions. Sediment thicknesses below the heavy black line show where positive density differences required for stable gravitational trapping are achieved.

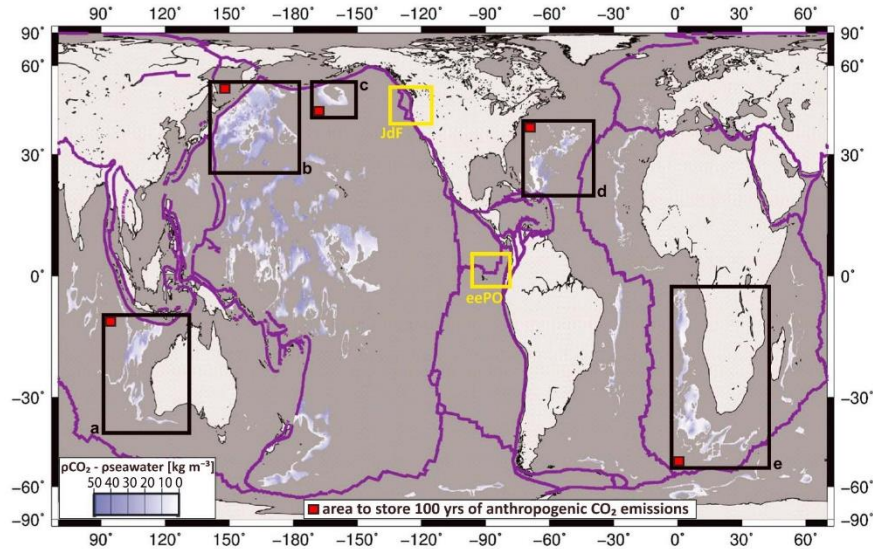
Technology (NIST) database (Linstrom, P. J., and W. G. Mallard (Eds.), NIST Chemistry WebBook, NIST Standard Reference Database Number 69, National Institute of Standards and Technology, Gaithersburg Md, 20899, retrieved November 12, 2012, <http://webbook.nist.gov>), which is based on the equation of state by *Span and Wagner* [1996]. Seawater density was estimated using the SeaWater MATLAB library [Fofonoff *et al.*, 1983], assuming a constant salinity of 35 psu [Brown *et al.*, 1995]. Figure 2 shows the density difference

between CO<sub>2</sub> and seawater as a function of temperature and pressure together with a phase diagram of carbon dioxide. Temperatures above 100°C are not considered because the density of CO<sub>2</sub> is too low to allow gravitational trapping.

[11] The density difference at the sediment-basement interface at each point in our 6' × 6' global grids is calculated using the estimated temperature and pressure. We combined this with the sediment thickness map to identify locations where (1) CO<sub>2</sub> is denser than seawater at the sediment-basement interface (Figure 2), and (2) the sediment thickness is between 200 m and 700 m (Figure 3). We choose a minimum thickness of 200 m to ensure a continuous low permeability blanket over minor basement topography such as fault ridges or seamounts that might puncture the sediment cover and allow the egress of basement fluids. To estimate the maximum sediment thickness, we have calculated the density difference for a wide range of lithospheric ages and sediment thicknesses using the GDH1 model for both water depth and heat flow, and assuming a hydrostatic sediment column (Figure 3). Based on global average conditions, GDH1 indicates a restricted zone where gravitational trapping is possible, and that anywhere with more than ~600 m of sediments CO<sub>2</sub> is likely to be gravitationally unstable due to the high temperatures. Using the HSCM (see supporting information “Figure S3”), the equivalent limit is ~1000 m. Hence, we settle on an upper sediment thickness limit of 700 m.

### 3. Discussions

[12] Much of the upper oceanic crust does not provide suitable locations for the geological sequestration of CO<sub>2</sub> by gravitational and physical trapping. Gravitational trapping ( $\rho_{\text{CO}_2} > \rho_{\text{seawater}}$ ) requires pressures higher than 25 MPa



**Figure 4.** An equal area map showing locations for stable geological sequestration of CO<sub>2</sub>. Shading shows the difference in density between CO<sub>2</sub> and seawater in areas where the sediment thickness is between 200 and 700 m and the CO<sub>2</sub> is denser than seawater. Five potential reservoirs (insets a–e) have been identified. The red box indicates the area required to store 100 yrs of current anthropogenic emissions of CO<sub>2</sub>, assuming a pillow lava thickness of 300 m and 10% porosity [Carlson and Herrick, 1990; Jarrard *et al.*, 2003; Johnson and Pruis, 2003]. Yellow boxes show regions in Figure 5.



**Table 1.** Properties of the Five Potential Reservoirs<sup>a</sup>

| Code | Location     | Area<br>[ $\times 10^6$ km <sup>2</sup> ] | Pore Volume<br>[ $\times 10^4$ km <sup>3</sup> ] | $\rho_{\text{CO}_2}$<br>[kg/m <sup>3</sup> ] | CO <sub>2</sub><br>[Gt] | $\Delta\rho$<br>[kg/m <sup>3</sup> ] | Age<br>[Myr] | Sed. Thickness<br>[m] | Distance<br>[km] |
|------|--------------|-------------------------------------------|--------------------------------------------------|----------------------------------------------|-------------------------|--------------------------------------|--------------|-----------------------|------------------|
| a    | Indian Ocean | 1.47                                      | 4.42                                             | 1066                                         | 47,162                  | 18                                   | 85           | 335                   | 1500             |
| b    | NW Pacific   | 3.97                                      | 11.9                                             | 1073                                         | 127,870                 | 24                                   | 100          | 310                   | 1300             |
| c    | S-Aleutians  | 0.43                                      | 1.30                                             | 1063                                         | 13,791                  | 15                                   | 60           | 275                   | 950              |
| d    | Bermuda      | 1.15                                      | 3.45                                             | 1066                                         | 36,780                  | 17                                   | 80           | 320                   | 1500             |
| e    | SE Atlantic  | 2.22                                      | 6.66                                             | 1062                                         | 70,701                  | 14                                   | 85           | 290                   | 1700             |

<sup>a</sup> $\Delta\rho$ , in situ excess density of CO<sub>2</sub> over seawater; age from [Müller et al., 2008]; sediment thickness from the NOAA database [Divins, 2003]. Distance of the reservoir from land is taken from the nearest stationary source of CO<sub>2</sub> according to the IEA GHG database [International Energy Agency Greenhouse Gas R&D Programme, 2002].

(~2500 m of water) and temperature between 0 and 30°C (Figure 2). The density of CO<sub>2</sub> decreases dramatically with decreasing pressure and increasing temperature, compared to a near constant density for seawater. The combination of high pressure and low temperature requires old ocean crust with relatively thin sediments. Note that within this  $p$ - $T$  window liquid CO<sub>2</sub> is the thermodynamically stable phase, with densities between 1040 and 1125 kg/m<sup>3</sup> compared with 140 kg/m<sup>3</sup> (at 100°C, 8 MPa) to 1045 kg/m<sup>3</sup> (at 30°C, 60 MPa) for supercritical CO<sub>2</sub>.

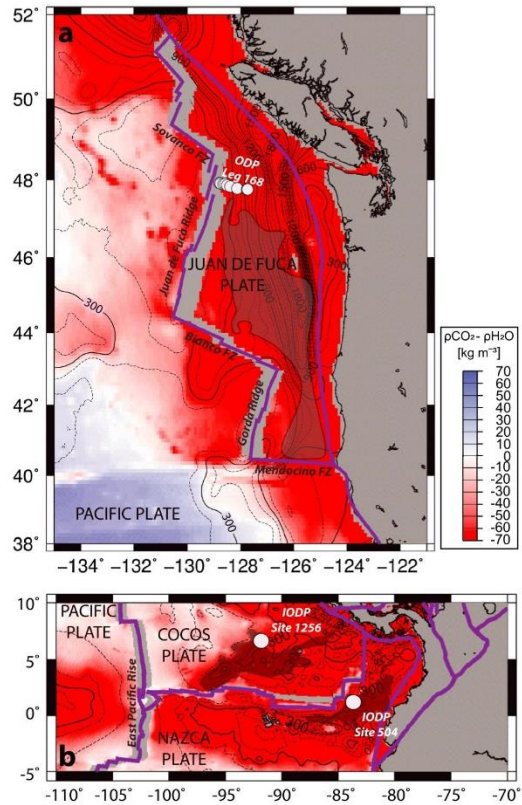
[13] Suitable reservoirs where physical (sediment between 200 and 700 m) and gravitational ( $\rho_{\text{CO}_2} > \rho_{\text{seawater}}$ ) trapping can be combined are shown on the global map (Figure 4); the global map produced using the HSCM is available in the supporting information “Figure S4”. Selected potential reservoirs are in the Indian Ocean between Indonesia and Australia (inset a); in the northwest Pacific Ocean near the east coast of Japan and Russia (inset b), and south of the Aleutian Islands (inset c); and in the Atlantic Ocean near Bermuda (inset d) and close to South Africa (inset e) (Table 1).

[14] We have identified these sites based on the positive  $\Delta\rho$  between CO<sub>2</sub> and seawater, the oceanic crustal age, the sediment thickness, and the distance to major industrial CO<sub>2</sub> sources [International Energy Agency Greenhouse Gas R&D Programme, 2002] (Table 1). Other areas also have suitable conditions for carbon dioxide trapping, but we have not yet explored these options due to their smaller sizes and lower  $\Delta\rho$ , although some are closer to land. We have computed the potential storage volume for each target (Table 1), assuming 300 m as reasonable thickness of permeable pillow lavas for old crust. Given an average porosity of 10% [Carlson and Herrick, 1990; Johnson and Pruett, 2003], even for old oceanic crust (e.g., ODP Hole 801C [Jarrard et al., 2003]), we determine the potential pore volume. The storage capacity in each area is between ~13,800 and 127,800 Gt of CO<sub>2</sub>. At the current global annual anthropogenic flux of ~35 Gt of CO<sub>2</sub> per year [Le Quere et al., 2009], even the smallest identified reservoir (inset c), could provide sufficient carbon dioxide sequestration capacity for several centuries (Figure 4).

[15] Contrary to previous suggestions [Goldberg et al., 2008; Slagle and Goldberg, 2011], sites on the Juan de Fuca Plate and in the eastern equatorial Pacific Ocean are unsuitable for gravitational trapping of carbon dioxide (Figure 5) because thick sediment covering young oceanic crust results in high temperatures at the sediment-basement interface, that renders CO<sub>2</sub> less dense than seawater.

[16] Our evaluation based on global data sets shows that CCS using subsea basalts as the storage medium has

considerable potential. However, regional investigations are needed to determine local sediment properties, thicknesses, continuity, and seafloor thermal gradients. Drilling to facilitate detailed lithological, physical, thermal, and hydrological characterization of the sediment overburden and target basalt formations is essential.



**Figure 5.** Map of density difference  $\Delta\rho$  between CO<sub>2</sub> and seawater at (a) Juan de Fuca Plate (with  $K_s = 2$  W/m/K) and (b) at eastern equatorial Pacific Ocean (with  $K_s = 1$  W/m/K). Sediment thicknesses are shown with black contour lines. The dark shadows show the previously suggested regions for deep-sea basalt CO<sub>2</sub> sequestration [Goldberg et al., 2008; Slagle and Goldberg, 2011].



#### 4. Conclusions

[17] Offshore basalt formations have been previously suggested as sites for geological carbon dioxide sequestration. We have used global data compilations to calculate the density of CO<sub>2</sub> in the pressure-temperature regime at the top of the basement throughout the world's oceans, and identified regions where CO<sub>2</sub> is denser than seawater. Previously suggested young sites on the eastern flank of the JdF Ridge and in the eastern equatorial Pacific (Sites 504, 1256) are not suitable for storing CO<sub>2</sub> because it is gravitationally unstable ( $\Delta\rho_{\text{CO}_2 - \text{seawater}} = 0$  to  $< -70 \text{ kg/m}^3$ ). However, we identify five large regions of old seabed where gravitational stability of stored CO<sub>2</sub> coincides with physical trapping by 200–700 m thickness of sediments. Using conservative assumptions about the porosity available, the smallest of these regions can store several centuries of anthropogenic CO<sub>2</sub> emissions.

[18] **Acknowledgments.** This research used data provided by the Integrated Ocean Drilling Program (IODP), the Ocean Drilling Program (ODP), and the Deep-Sea Drilling Project (DSDP). Funding for this research was provided by the University of Southampton Vice Chancellor's Scholarship to C. M., and NERC grant NE/I006311/1 to D.A.H.T. We thank Juerg M. Matter, Nick Woodman, and Martin R. Palmer for their discussions that assisted in the development of these ideas. We thank the Editor Peter Stratton, Robert Harris and a second anonymous reviewer for their constructive comments that have improved the paper.

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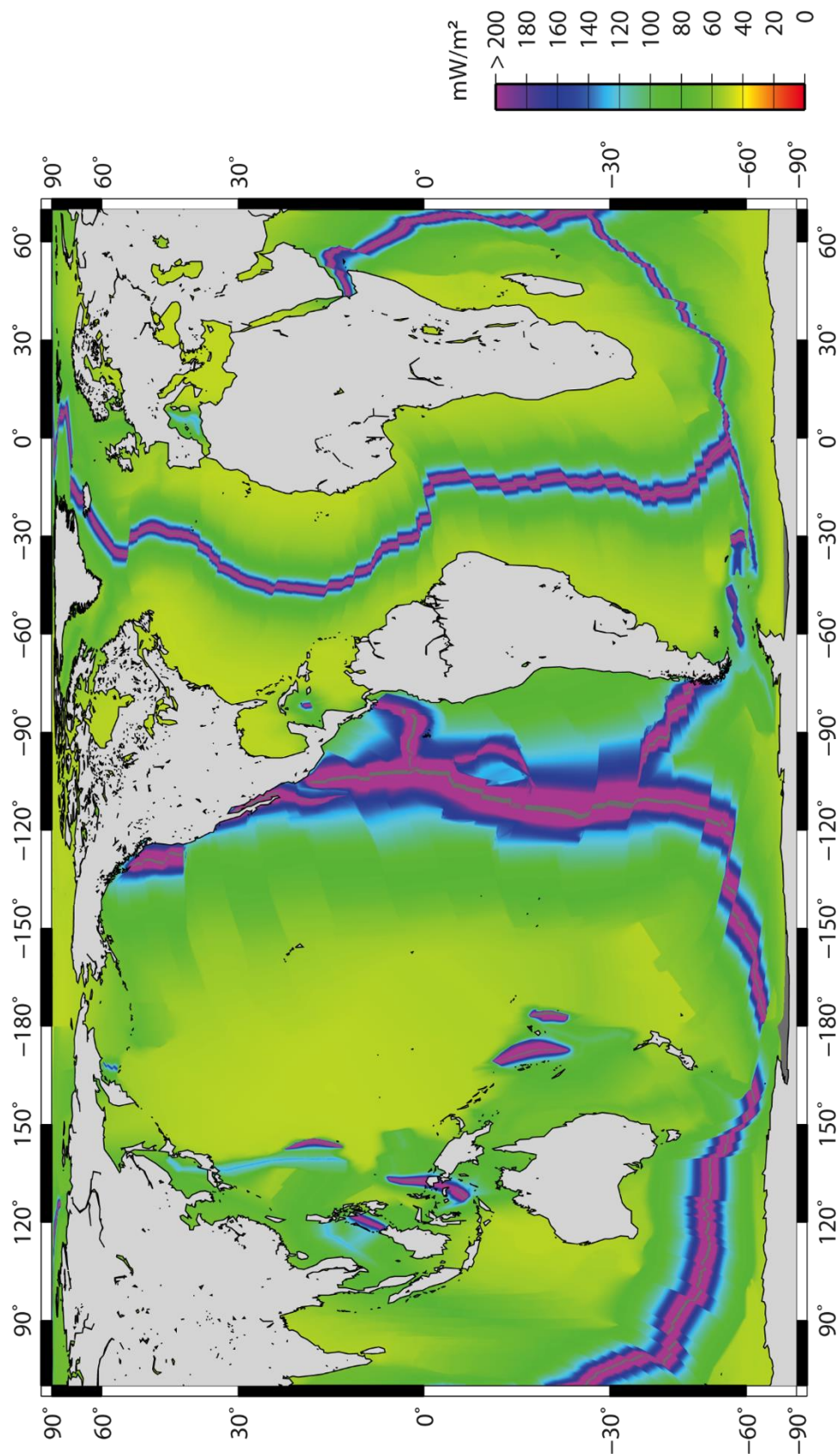
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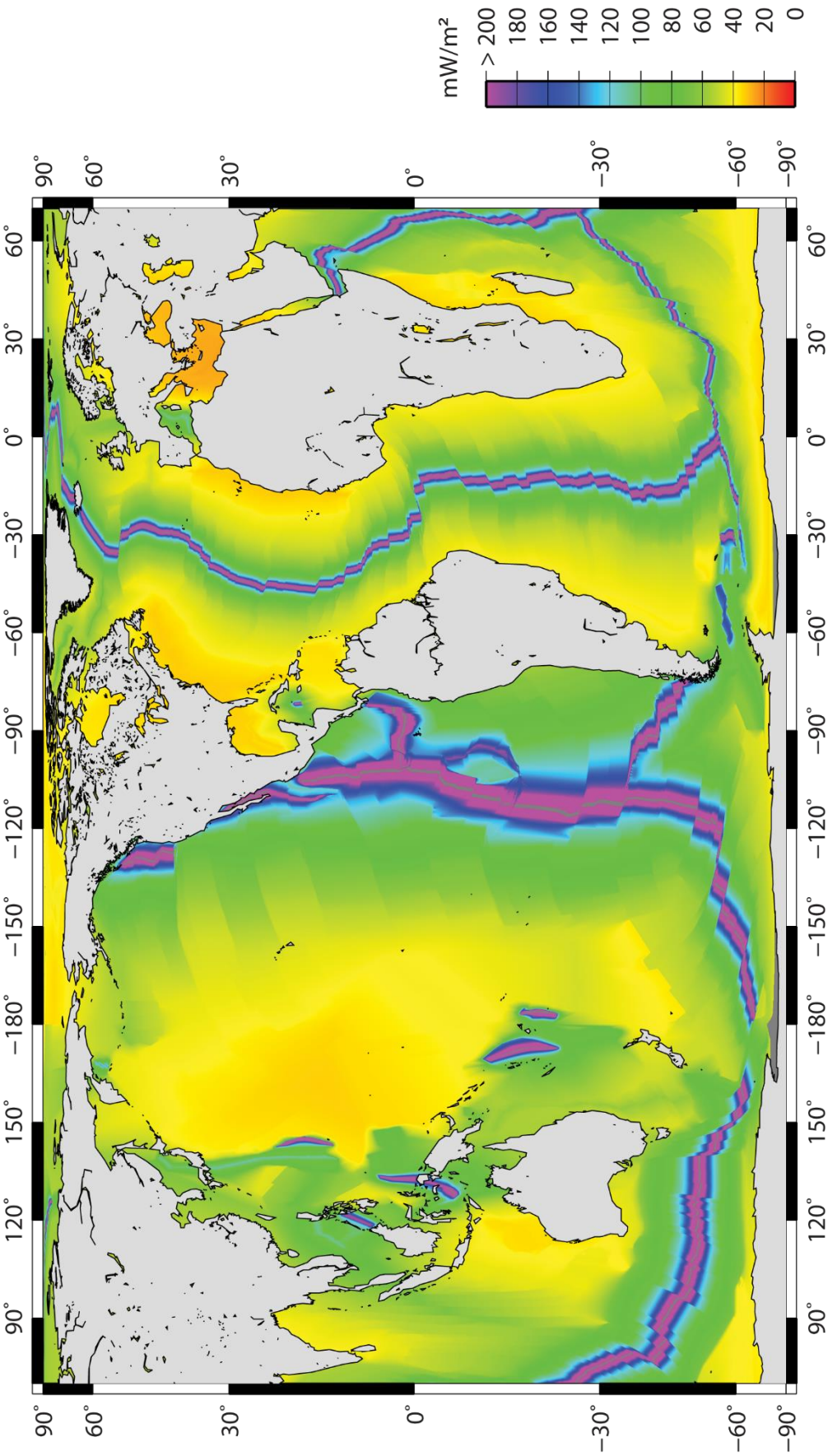
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## A-Chapter 2

CH2-1A: Heat flow map generated with the GDH1 model.

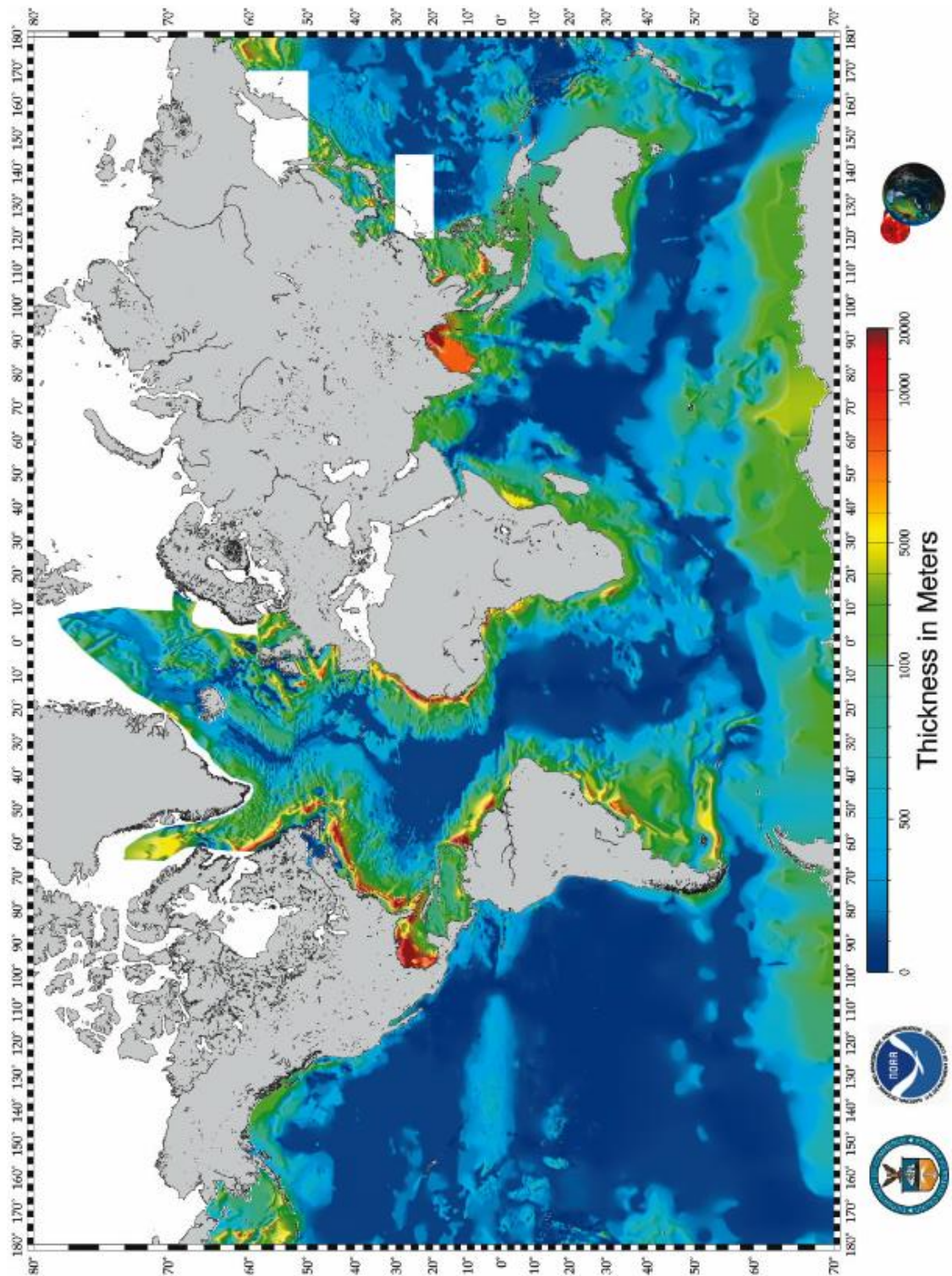


**CH2-1B:** Heat flow map generated with the HSCM.

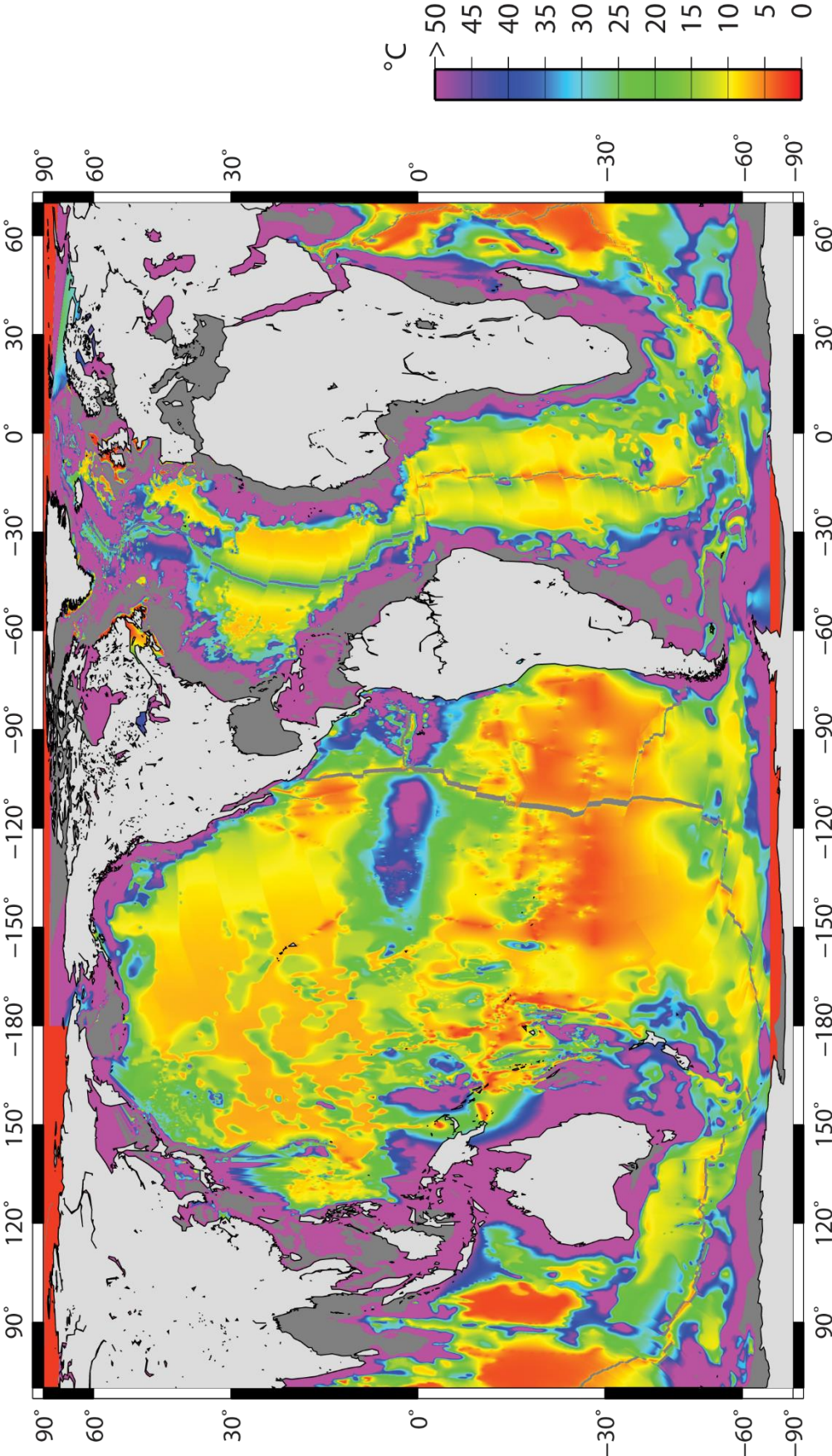




**CH2-2:** NOAA National Geophysical Data Center (NGDC) sediment thickness map [Divins, 2003]. The data were gridded with a grid spacing of 5 arc-minutes by 5 arc-minutes.

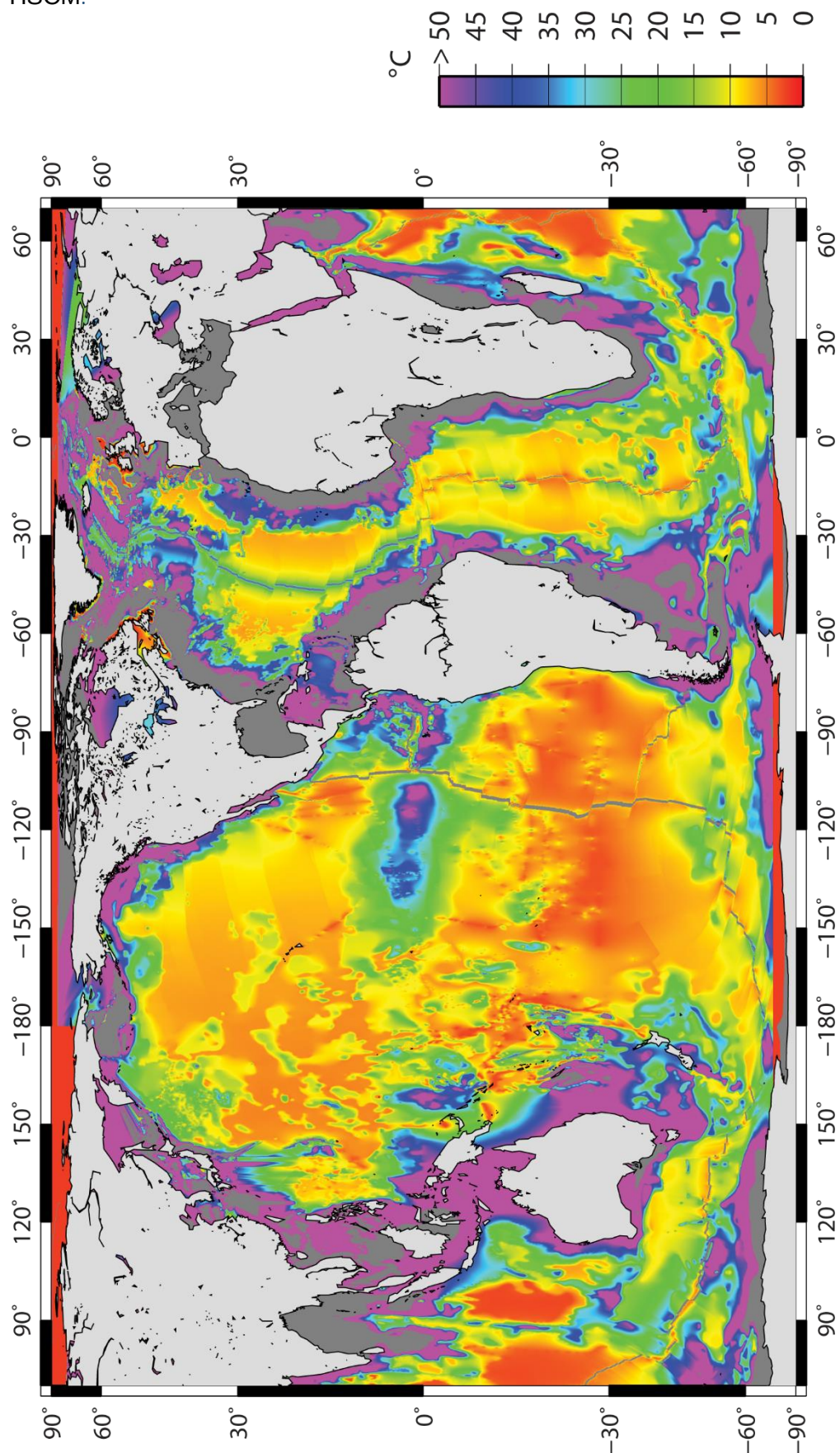


**CH2-3A:** Map of temperature at the sediment-basement interface, generated with the GDH1 model.



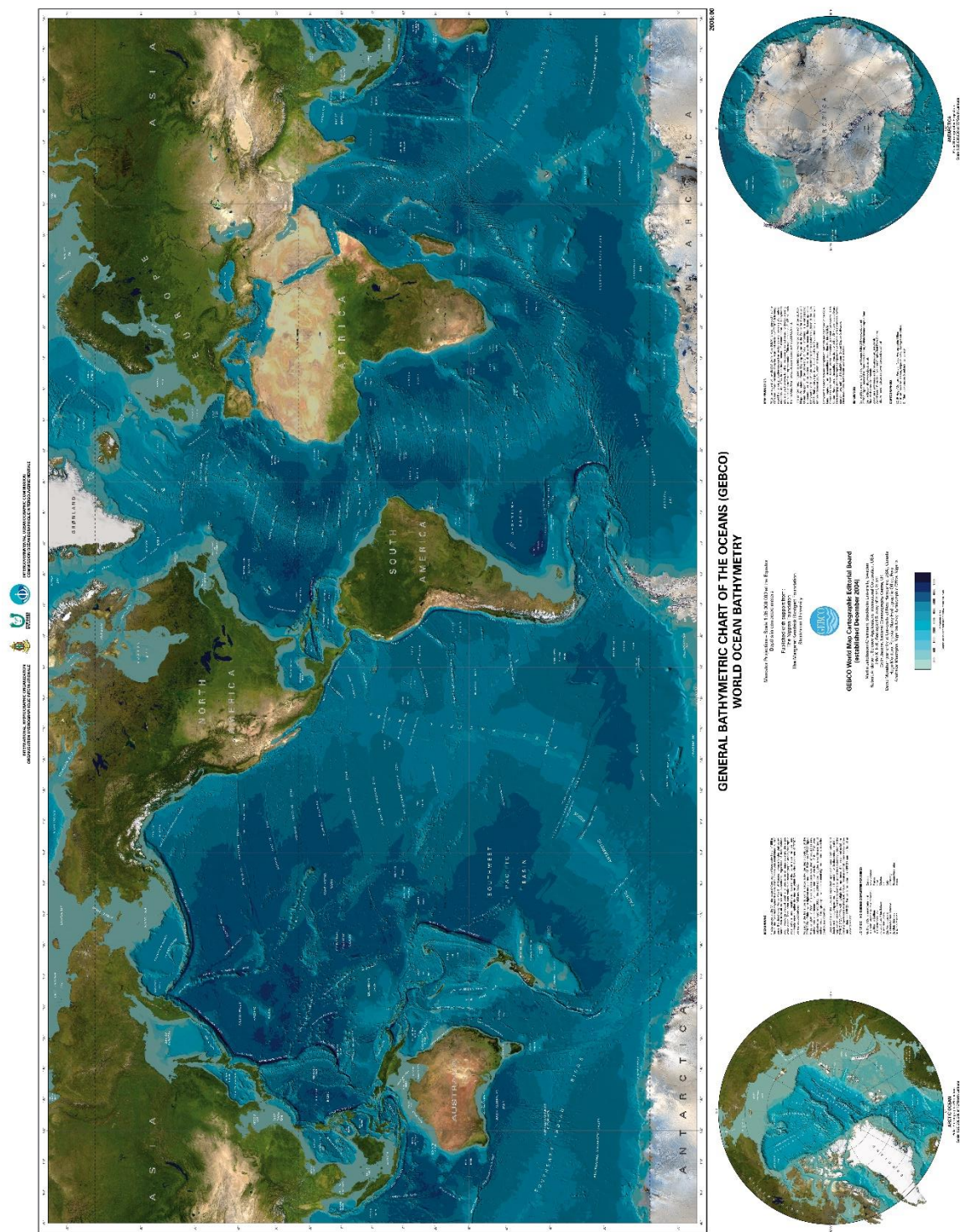


**CH2-3B:** Map of temperature at the sediment-basement interface, generated with the HSCM.



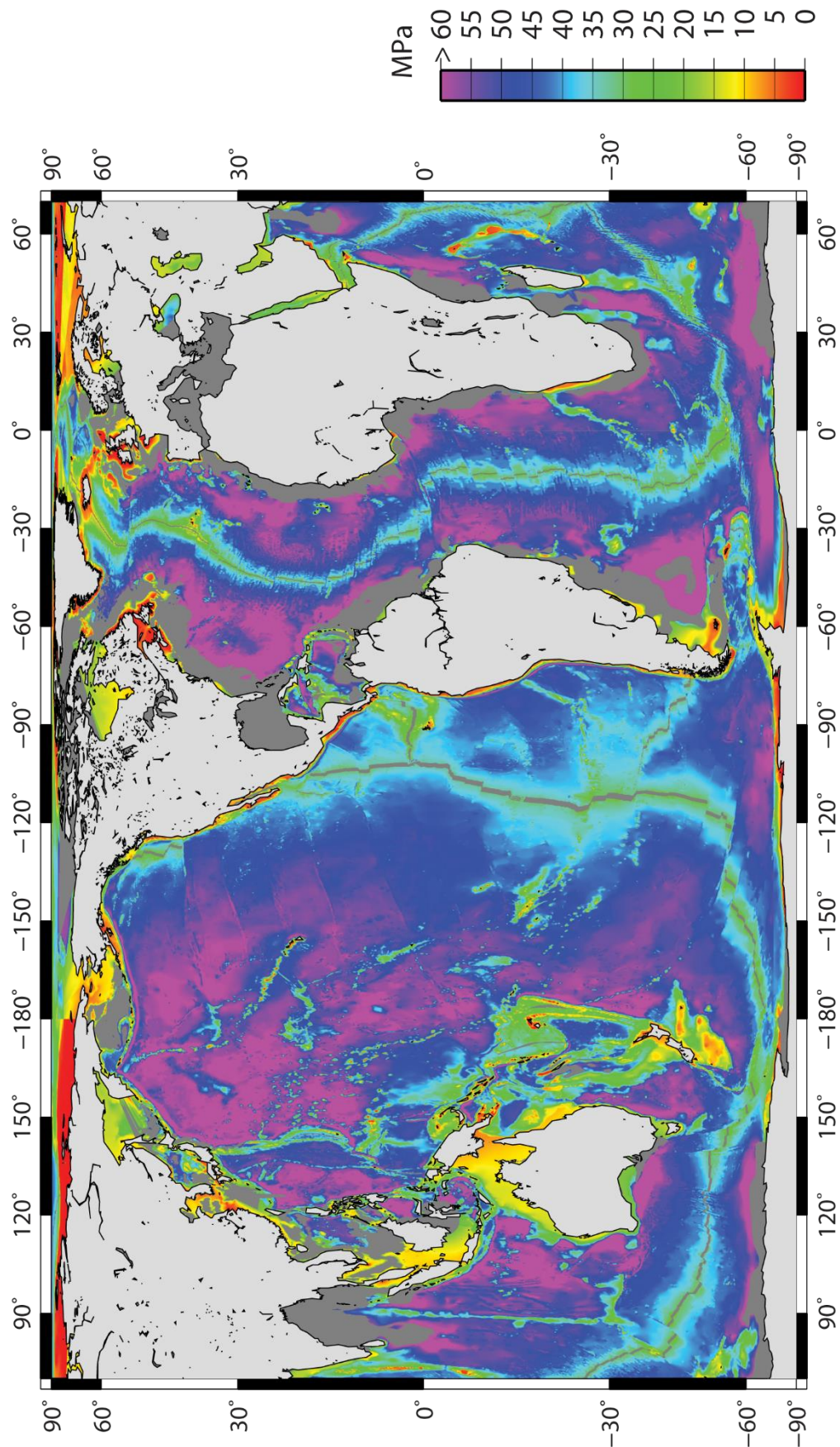


**CH2-4: General Bathymetric Chart of the Oceans (GEBCO) – gridfive world bathymetry map [IOC et al., 2003].**

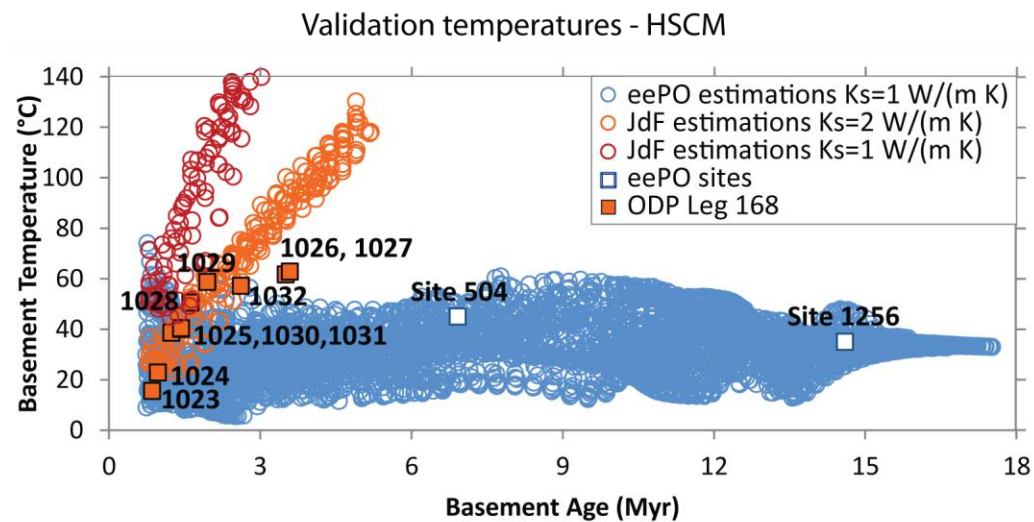
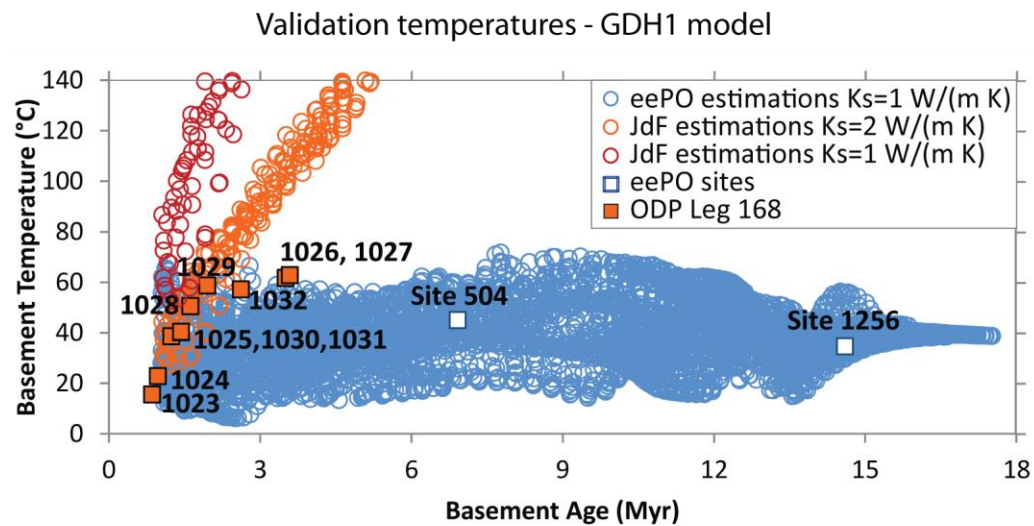




**CH2-5:** Map of hydrostatic pressure at the sediment-basement interface.

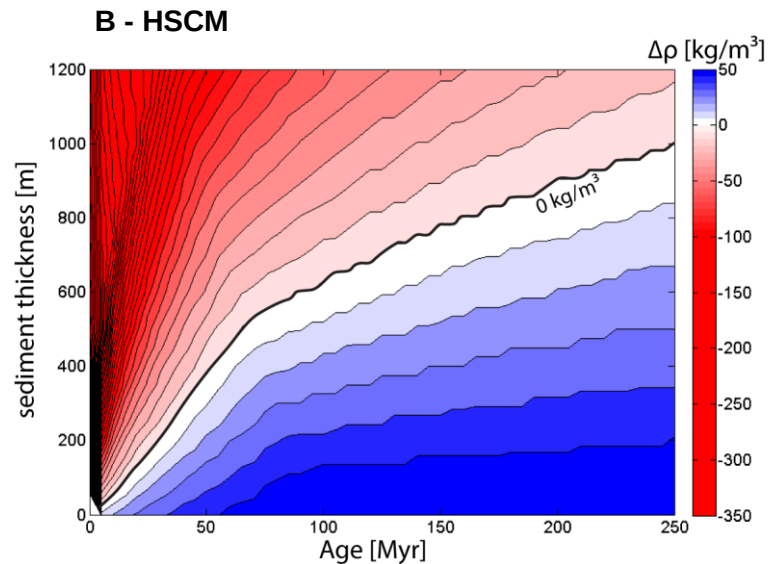
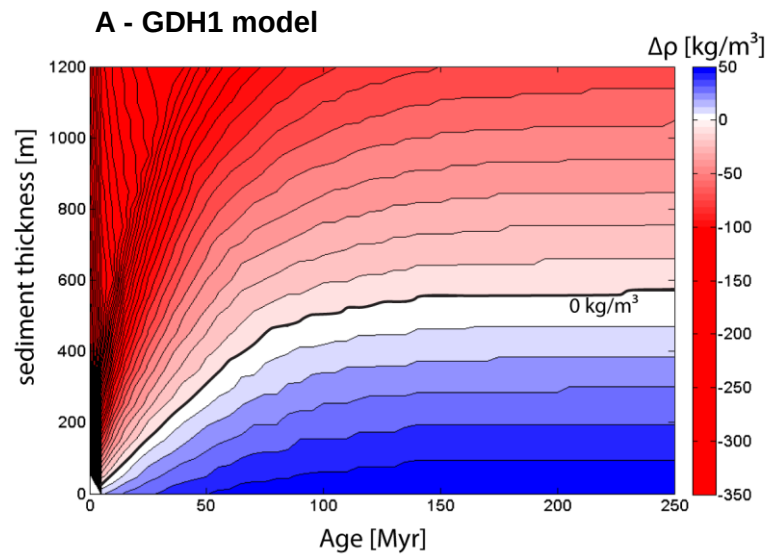


**CH2-6:** Comparison between estimated temperatures (from GDH1model at the top, and from HSCM at the bottom) in the eastern equatorial Pacific Ocean (eePO) and the Juan de Fuca Plate (JdFP), and measured downhole temperatures at the sediment-basement interface [Marieni *et al.*, 2013]. White squares: data from eePO [Alt *et al.*, 1993; Teagle *et al.*, 2006]; orange squares: data from JdFP [Davis *et al.*, 1997]. Circles: estimated values in the eePO (blue), and on the JdFP (red with  $K_s = 1$  W/(m K); orange with  $K_s = 2$  W/(m K)).



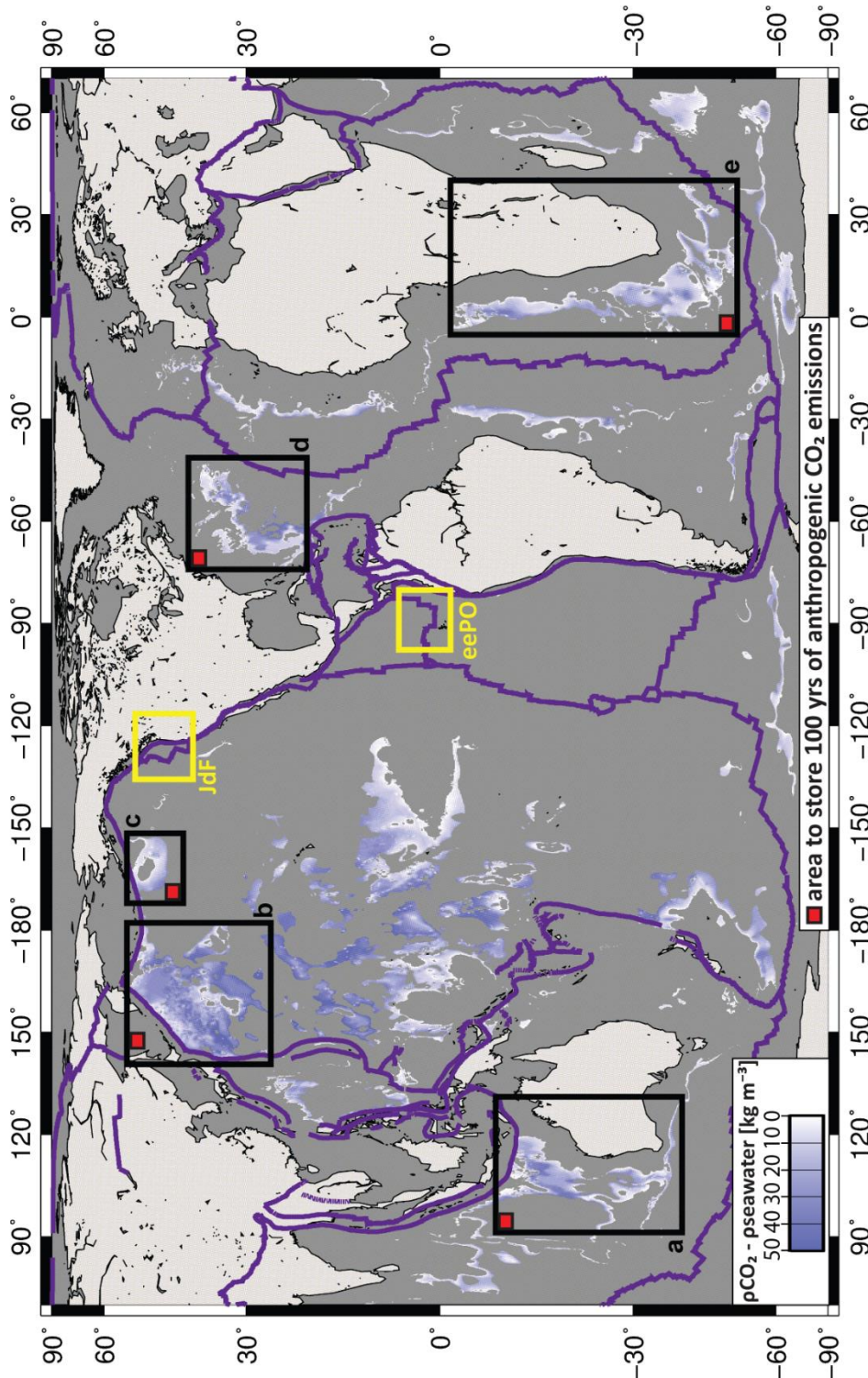
## A-Chapter 3

**CH3-1:** Density difference between CO<sub>2</sub> and seawater at the sediment-basement interface as a function of plate age and sediment thickness using **A-** GDH1 and **B-** HSCM models to determine both water depth and thermal conditions. Sediment thicknesses below the heavy black line show where positive density differences required for stable gravitational trapping are achieved.

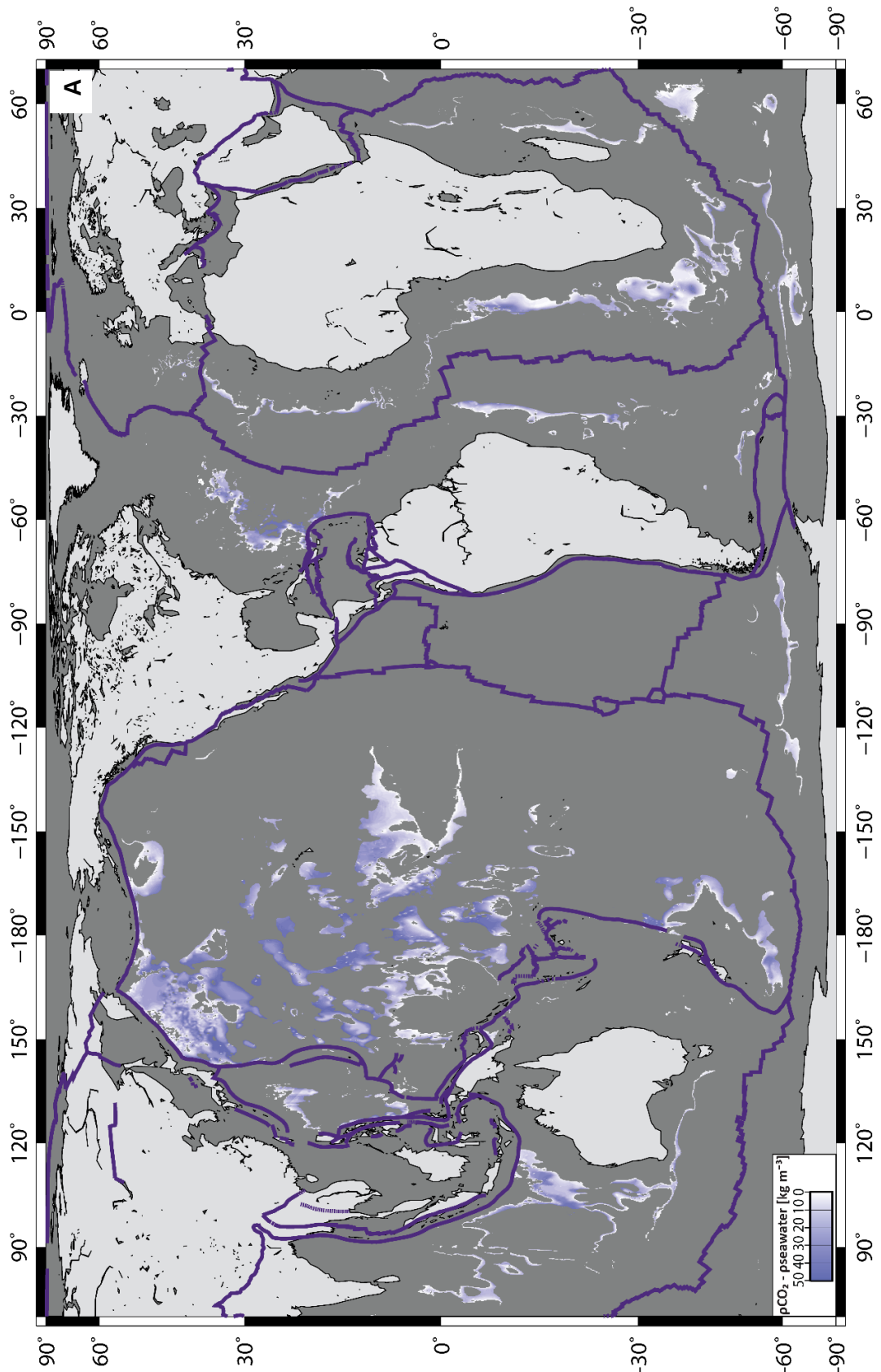




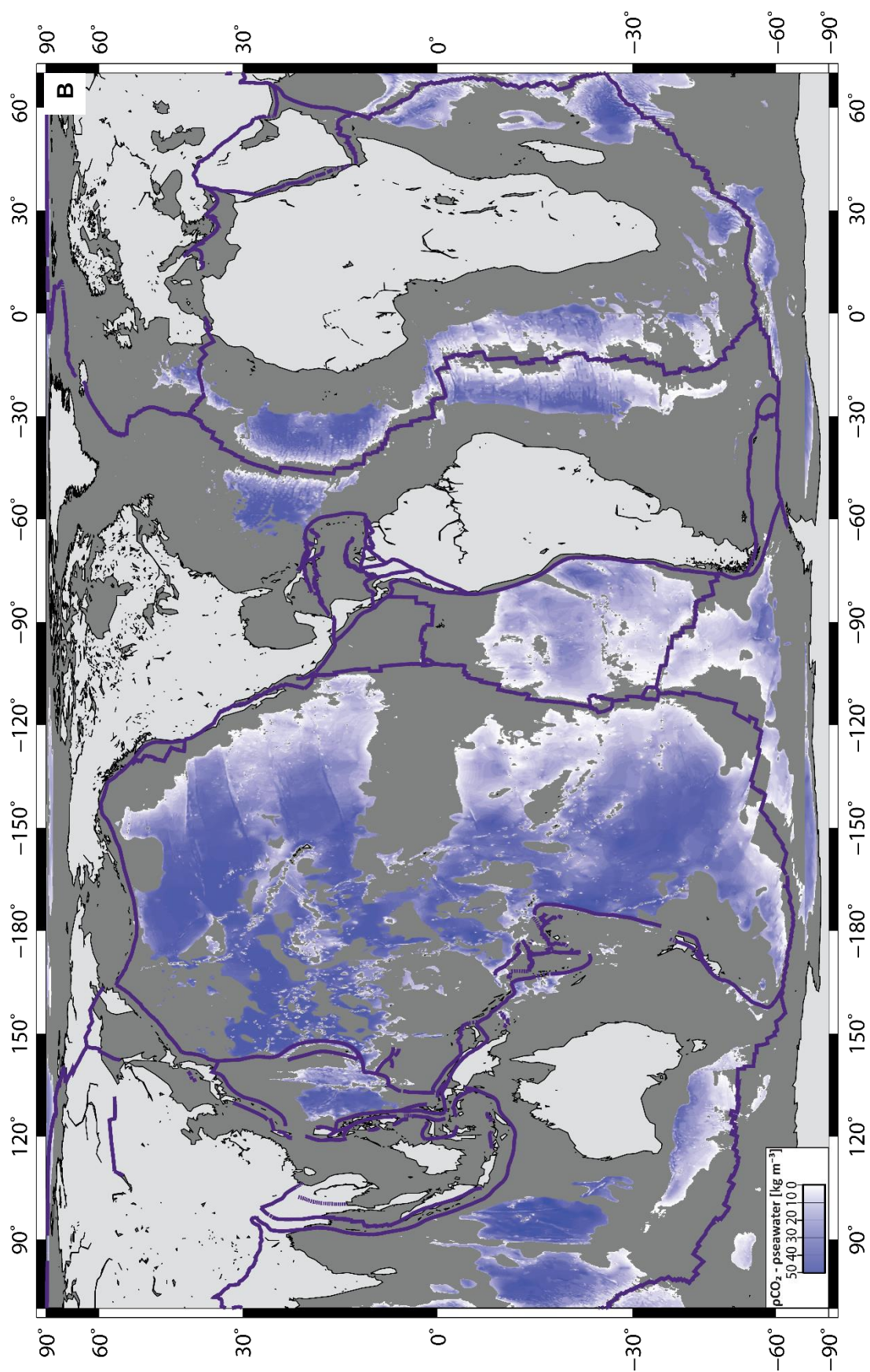
**CH3-2:** An equal area map showing locations for stable geological sequestration of CO<sub>2</sub> using the HSCM. Shading shows the difference in density between CO<sub>2</sub> and seawater in areas where the sediment thickness is between 200 and 700 m and the CO<sub>2</sub> is denser than seawater. Five potential reservoirs (**a/b/c/d/e**) have been identified. The red box indicates the area required to store 100 yrs of current anthropogenic emissions of CO<sub>2</sub>, assuming a pillow lava thickness of 300 m and 10% porosity [Carlson and Herrick, 1990; Jarrard et al., 2003; Johnson and Pruis, 2003]. Yellow boxes show the eePO and JdF regions.

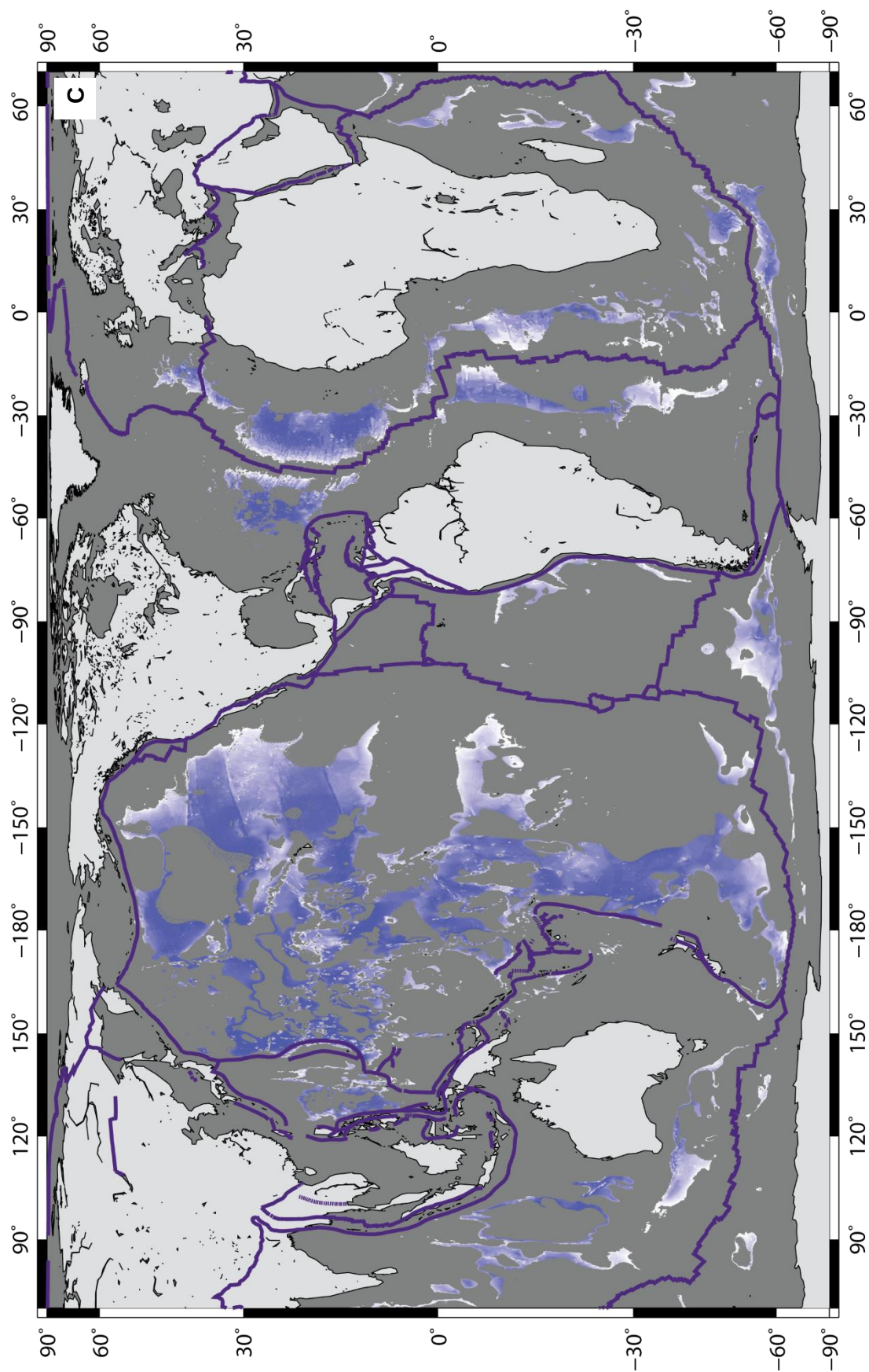


**CH3-3:** An equal area map showing locations for stable geological sequestration of CO<sub>2</sub> using the GDH1 model. Shading shows the difference in density between CO<sub>2</sub> and seawater in areas where CO<sub>2</sub> is denser than seawater, and the sediment thickness is **A**- between 200 and 500 m; **B**- between 0 and 200 m; **C**- between 100 and 200 m; **D**- between 0 and 500 m; and **E**- between 0 and 700 m.

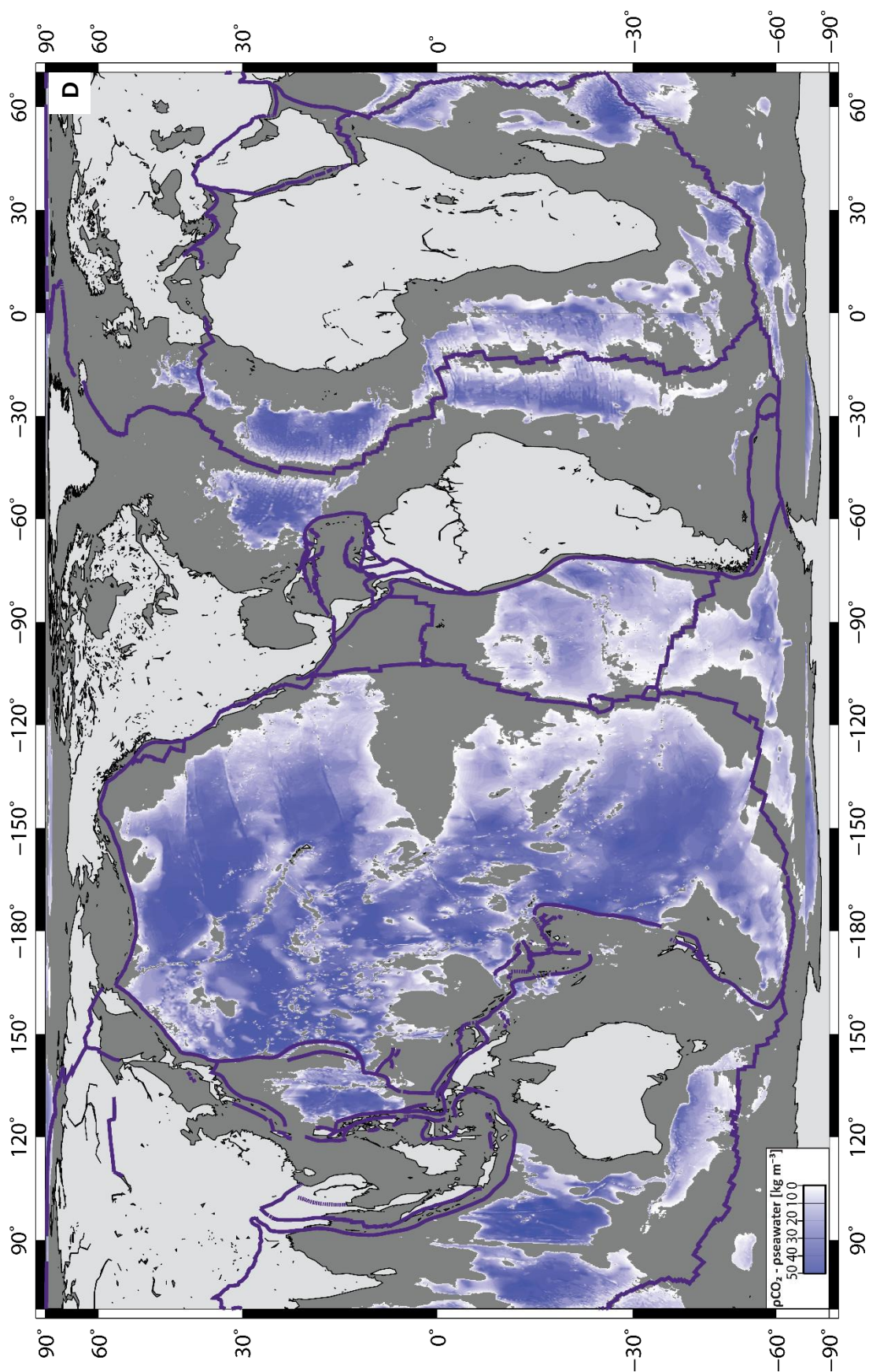




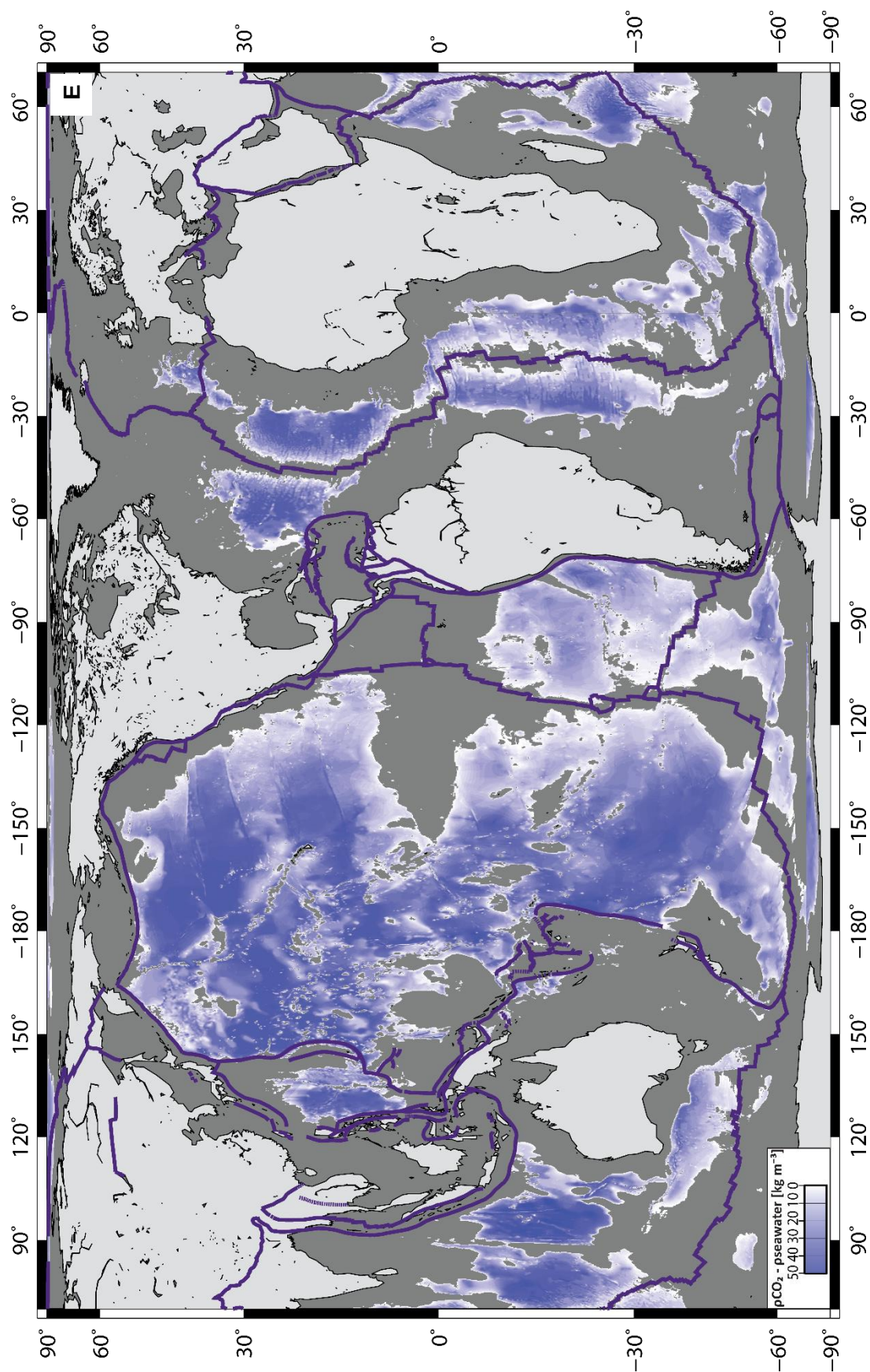












## A-Chapter 4, Methods

**CH4-1:** Alkalinity titration for seawater, modified from *Haraldsson et al. (1997)*.

### MATERIAL:

Syringe (0.1-1 mL)

Becker 10mL

0.0005M HCl (in dark glass bottle) – solution prepared from HCl 0.1 mol/L (0.1M) ConvoL

Deionized (DI) water

pH electrode (FB68801, Fisherbrand™)

pH meter (Fisher Scientific™ accumet™ AB15/15+)

Magnetic stirrer (Metrohm 728), 1.5 cm magnetic follower (Fisherbrand™) and retriever

Titration (Metrohm 775 Dosimat)

### CALIBRATION:

Calibrate the pH probe using mercury free buffer solutions Reagecon at pH 4.00, 7.00 and 10.00. Rinse the pH probe with DI water.

### TITRATION:

Inject 0.5 mL of sample in 4.5 mL of DI water (necessary to have enough “depth” for the pH measurement). Insert the magnetic follower and power the magnetic stirrer. Run the titration. Always collect the magnetic follower with the retriever before rinsing the equipment. Always rinse the equipment between titrations.

### CALCULATIONS:

Calculate the Gran function (G):

$$G = (V + V_0) * 10^{-pH} \quad (19)$$

where V = the volume of acid added and  $V_0$  = the initial volume of solution (0.5mL in this case).

Plot G as a function of the total volume of acid added; find the equivalence point by extrapolating the linear part of the graph to its intersect with the x-axis.

The total alkalinity of the sample (in units of mmol/L) is given by:

$$\frac{\text{Volume of acid at equivalence point (mL)} * \text{acid molarity (mol/L)}}{V_0 \text{ (mL)}} * 1000 \quad (20)$$

**CH4-2:** Methodology for measuring the cation concentrations in seawater with inductively coupled plasma optical emission spectrometry (ICP-OES, Perkin-Elmer Optima 4300 DV).

**MATERIAL:**

“A” x 20 mL acid clean (10% HNO<sub>3</sub>) bottles (depending on “A” samples)  
 8 x 125 mL acid clean (10% HNO<sub>3</sub>) bottles  
 3 x 20 mL acid clean (10% HNO<sub>3</sub>) bottles – for intermediate bottles  
 3% HNO<sub>3</sub>, 5ppb In, 5ppb Re, 20ppb Be (for diluting samples and standards)  
 10% HNO<sub>3</sub>  
 Pipettes 100-1000 µL and 10-100 µL  
 Deionized (DI) water (for cleaning pipette)  
 6M HCl (for cleaning pipette)

**INITIAL CONCENTRATIONS of elements (ppm):**

**K<sup>+</sup>** = 9777.6  
**Ca<sup>2+</sup>** = 9631.4  
**S** = 10000  
**Mg<sup>2+</sup>** = 9521.8  
**Na<sup>+</sup>** = 9688.3  
**Sr<sup>2+</sup>** = 999  
**Si** = 9552.6

**CALCULATION FOR DILUTIONS:**

$$C_1 = (C_0 * V_0)/V_1$$

**C<sub>0</sub>** Initial concentrations of a certain element [ppm = mg/kg]  
**C<sub>1</sub>** Final concentrations of a certain element [ppm = mg/kg]  
**V<sub>0</sub>** Initial volume of a certain element [mL]  
**V<sub>1</sub>** Final volume of a certain element (diluted with 3% HNO<sub>3</sub>, 5ppb In, 5ppb Re, 20ppb Be) [mL]

**Sample dilution:**

50 fold dilution (0.15 mL of sample in 7.35 mL of 3% HNO<sub>3</sub> (5ppb In, 5ppb Re, 20ppb Be)).

**Standards dilutions:**

**V<sub>1</sub>** = 100 mL

We are making 6 X standards (multiple elements) each of 100 mL with the following concentrations **C<sub>1</sub>** (in ppm):

|                        | <b>Std 1</b> | <b>Std 2</b> | <b>Std 3</b> | <b>Std 4</b> | <b>Std 5</b> | <b>Std 6</b> |
|------------------------|--------------|--------------|--------------|--------------|--------------|--------------|
| <b>K<sup>+</sup></b>   | 1            | 5            | 7            | 8            | 10           | 15           |
| <b>Ca<sup>2+</sup></b> | 1            | 5            | 7            | 8            | 10           | 15           |
| <b>S</b>               | 5            | 10           | 15           | 18           | 20           | 25           |
| <b>Mg<sup>2+</sup></b> | 18           | 22           | 25           | 27           | 30           | 32           |
| <b>Na<sup>+</sup></b>  | 190          | 200          | 210          | 215          | 220          | 225          |
| <b>Sr<sup>2+</sup></b> | 0.05         | 0.15         | 0.20         | 0.50         | 1.00         | 2.00         |
| <b>Si</b>              | 0.05         | 0.10         | 0.50         | 1.00         | 1.50         | 2.00         |

#### CALIBRATIONS:

- Using “blank samples”, with only 3% HNO<sub>3</sub> (5ppb In, 5ppb Re, 20ppb Be) or 3% HNO<sub>3</sub>
- Using 6 standards
- Using one of the standard as a drift monitor
- Using IAPSO
- Using seawater from tank CaribSea Inc (not acidified with CO<sub>2</sub>)

IAPSO Reference [Summerhayes and Thorpe, 1996]:

|     | Ca <sup>2+</sup> | K <sup>+</sup> | Mg <sup>2+</sup> | Na <sup>+</sup> | S (total) | Si  | Sr <sup>2+</sup> |
|-----|------------------|----------------|------------------|-----------------|-----------|-----|------------------|
| ppm | 412              | 399            | 1290             | 10770           | 904       | 2.8 | 7.9              |

SEAWATER Reference [Millero et al., 2008]:

|     | Ca <sup>2+</sup> | K <sup>+</sup> | Mg <sup>2+</sup> | Na <sup>+</sup> | SO <sub>4</sub> <sup>2-</sup> |  | Sr <sup>2+</sup> |
|-----|------------------|----------------|------------------|-----------------|-------------------------------|--|------------------|
| ppm | 412              | 399            | 1284             | 10781           | 2712                          |  | 8                |

#### **NB**

- Preserve major elements in samples with a drop of 3% HNO<sub>3</sub> in 0.5 mL of sample.
- Preserve the samples in vial closed with parafilm and refrigerated to avoid evaporation effects.
- For the analysis we used 4-5 mL of each 50 fold diluted samples

#### ACCURACY and precision on IAPSO:

|                                                                                   |        | Ca <sup>2+</sup> | K <sup>+</sup> | Mg <sup>2+</sup> | Na <sup>+</sup> | S (total) | Si   | Sr <sup>2+</sup> |
|-----------------------------------------------------------------------------------|--------|------------------|----------------|------------------|-----------------|-----------|------|------------------|
| <b>Average N=9</b>                                                                | ppm    | 420              | 393            | 1280             | 10800           | 900       | 3.02 | 7.91             |
| <b>Accuracy</b>                                                                   | %      | 1.84             | 1.58           | 0.87             | 0.04            | 0.45      | 7.91 | 0.08             |
| <b>Precision</b>                                                                  | %RSD   | 1.85             | 2.61           | 1.77             | 1.26            | 0.38      | 55.5 | 1.09             |
| <b>Error 1σ</b>                                                                   | ppm    | 7.75             | 10.3           | 22.6             | 136             | 3.43      | 1.68 | 0.09             |
|                                                                                   | mmol/L | 0.19             | 0.26           | 0.93             | 5.92            | 0.11      | 0.06 | 0.001            |
| <b>Error 1σ<br/>in logarithmic unit<br/>σ<sub>log</sub> = 0.434 * (σ/average)</b> |        | 0.01             | 0.01           | 0.01             | 0.01            | 0.002     | 0.24 | 0.005            |

ACCURACY on standards (ppm):

|                         |             | K <sup>+</sup> | Ca <sup>2+</sup> | S    | Mg <sup>2+</sup> | Na <sup>+</sup> | Sr <sup>2+</sup> | Si   |
|-------------------------|-------------|----------------|------------------|------|------------------|-----------------|------------------|------|
| <b>Std 1</b>            | known value | 0.95           | 0.94             | 4.86 | 18.2             | 189             | 0.05             | 0.03 |
|                         | average N=9 | 0.99           | 1.01             | 4.84 | 18.4             | 190             | 0.05             | 0.03 |
| <b>Std 2</b>            | known value | 5.09           | 5.06             | 9.58 | 22.7             | 203             | 0.11             | 0.09 |
|                         | average N=9 | 5.18           | 5.28             | 9.66 | 22.9             | 202             | 0.12             | 0.09 |
| <b>Std 3</b>            | known value | 7.25           | 7.21             | 15.2 | 24.7             | 217             | 0.18             | 0.47 |
|                         | average N=9 | 7.45           | 7.55             | 15.2 | 24.8             | 217             | 0.19             | 0.48 |
| <b>Std 4</b>            | known value | 8.21           | 8.16             | 18.2 | 26.4             | 219             | 0.49             | 0.94 |
|                         | average N=9 | 8.46           | 8.46             | 18.3 | 26.7             | 219             | 0.50             | 0.95 |
| <b>Std 5</b>            | known value | 9.93           | 9.88             | 20.2 | 30.4             | 224             | 0.98             | 1.40 |
|                         | average N=9 | 9.98           | 10.0             | 20.1 | 30.1             | 223             | 0.98             | 1.39 |
| <b>Std 6</b>            | known value | 15.4           | 15.3             | 24.8 | 32.1             | 227             | 1.94             | 1.89 |
|                         | average N=9 | 15.1           | 14.8             | 24.8 | 31.7             | 228             | 1.94             | 1.89 |
| <b>Average accuracy</b> |             |                |                  |      |                  |                 |                  |      |
|                         | %           | 1.34           | 3.52             | 1.15 | 1.00             | 0.26            | 6.35             | 1.37 |

PRECISION on standards used as drift monitors:

|                  | DM-run 2<br>Std 3 | DM-run 3<br>Std 2 | DM-run 4<br>Std 5 | Average |
|------------------|-------------------|-------------------|-------------------|---------|
|                  | %RSD              | %RSD              | %RSD              | %RSD    |
| K <sup>+</sup>   | 1.09              | 0.93              | 0.91              | 0.98    |
| Ca <sup>2+</sup> | 1.64              | 1.10              | 0.94              | 1.23    |
| S                | 0.18              | 0.20              | 0.21              | 0.20    |
| Mg <sup>2+</sup> | 0.73              | 0.46              | 0.37              | 0.52    |
| Na <sup>+</sup>  | 0.82              | 0.62              | 0.87              | 0.77    |
| Sr <sup>2+</sup> | 0.43              | 0.55              | 0.86              | 0.61    |
| Si               | 0.71              | 1.30              | 0.60              | 0.87    |

**CH4-3:** Methodology for measuring the anion concentrations in seawater with ion chromatography (Dionex ICS2500).

**MATERIAL:**

“A” x 20 mL bottles (depending on “A” samples)  
8 x 125 mL bottles  
Pipettes 100-1000 µL  
Deionized (DI) water (for diluting samples and standards)  
Deionized (DI) water (for cleaning pipette)  
6M HCl (for cleaning pipette)

**INITIAL CONCENTRATIONS of elements (ppm):**

**Cl<sup>-</sup>** = 5028.4  
**SO<sub>4</sub><sup>2-</sup>** = 1000  
**Br<sup>-</sup>** = 1000  
**F<sup>-</sup>** = 1000

**CALCULATION FOR DILUTIONS:**

$$C_1 = (C_0 * V_0)/V_1$$

$C_0$  Initial concentrations of a certain element [ppm = mg/kg]  
 $C_1$  Final concentrations of a certain element [ppm = mg/kg]  
 $V_0$  Initial volume of a certain element [ml]  
 $V_1$  Final volume of a certain element (diluted with DI water) [mL]

**Sample dilution:**

100 fold dilution (0.10 mL of sample in 9.90 mL of DI water).

**Standards dilutions:**

$V_1$  = 50 mL for Std 1, and 2  
 $V_1$  = 100 mL for Std 3, 4, 5, and 6

We are making 6 X standards (multiple elements) with the following concentrations  $C_1$  (in ppm):

|                                    | <b>Std 1</b> | <b>Std 2</b> | <b>Std 3</b> | <b>Std 4</b> | <b>Std 5</b> | <b>Std 6</b> |
|------------------------------------|--------------|--------------|--------------|--------------|--------------|--------------|
| <b>Cl<sup>-</sup></b>              | 3.24         | 32.4         | 108          | 180          | 300          | 500          |
| <b>SO<sub>4</sub><sup>2-</sup></b> | 0.97         | 9.72         | 32.4         | 54.0         | 90.0         | 150          |
| <b>Br<sup>-</sup></b>              | 0.03         | 0.32         | 1.08         | 1.80         | 3.00         | 5.00         |
| <b>F<sup>-</sup></b>               | 0.01         | 0.13         | 0.43         | 0.72         | 1.20         | 2.00         |

**CALIBRATIONS:**

- Using “blank samples”, with only DI water
- Using 6 standards
- Using one of the standard as a drift monitor
- Using IAPSO
- Using seawater from tank CaribSea Inc (not acidified with CO<sub>2</sub>)

IAPSO Reference [Summerhayes and Thorpe, 1996]:

|     | Cl <sup>-</sup> | SO <sub>4</sub> <sup>2-</sup> | Br <sup>-</sup> | F <sup>-</sup> |
|-----|-----------------|-------------------------------|-----------------|----------------|
| ppm | 19354           | 2712                          | 67              | 1.3            |

SEAWATER Reference [Millero et al., 2008]:

|     | Cl <sup>-</sup> | SO <sub>4</sub> <sup>2-</sup> | Br <sup>-</sup> | F <sup>-</sup> |
|-----|-----------------|-------------------------------|-----------------|----------------|
| ppm | 19353           | 2712                          | 67              | 1              |

#### NB

- Preserve the samples in vial closed with parafilm and refrigerated to avoid evaporation effects.
- For the analysis we used 5 mL of each 100 fold diluted samples

ACCURACY and precision on IAPSO:

|             |      | Cl <sup>-</sup> | SO <sub>4</sub> <sup>2-</sup> | Br <sup>-</sup> | F <sup>-</sup> |
|-------------|------|-----------------|-------------------------------|-----------------|----------------|
| Average N=5 | ppm  | 19700           | 2200                          | 66.4            | 1.02           |
| Accuracy    | %    | 1.58            | 2.65                          | 17.67           | 21.84          |
| Precision   | %RSD | 0.70            | 1.16                          | 25.73           | 59.13          |

ACCURACY on standards (ppm):

|                                  |             | Cl <sup>-</sup> | SO <sub>4</sub> <sup>2-</sup> | Br <sup>-</sup> | F <sup>-</sup> |
|----------------------------------|-------------|-----------------|-------------------------------|-----------------|----------------|
| Std 1                            | known value | 3.26            | 0.95                          | 0.03            | 0.01           |
|                                  | average N=6 | 4.82            | 0.80                          | 0.04            | 0.01           |
| Std 2                            | known value | 33.2            | 9.63                          | 0.319           | 0.13           |
|                                  | average N=6 | 31.8            | 8.35                          | 0.321           | 0.13           |
| Std 3                            | known value | 111             | 32.2                          | 1.06            | 0.42           |
|                                  | average N=6 | 110             | 30.7                          | 1.06            | 0.40           |
| Std 4                            | known value | 185             | 53.6                          | 1.77            | 0.69           |
|                                  | average N=6 | 185             | 53.2                          | 1.71            | 0.67           |
| Std 5                            | known value | 307             | 89.0                          | 2.95            | 1.15           |
|                                  | average N=6 | 308             | 90.5                          | 3.01            | 1.18           |
| Std 6                            | known value | 514             | 149                           | 4.93            | 1.93           |
|                                  | average N=6 | 514             | 149                           | 4.92            | 1.92           |
| Average accuracy %               |             | 8.83            | 5.97                          | 4.35            | 2.63           |
| Average accuracy without Std 1 % |             | 1.04            | 4.12                          | 1.39            | 1.85           |

PRECISION on standards used as drift monitors:

|                               | DM-run 2<br>Std 4 | DM-run 3<br>Std 3 | DM-run 4<br>Std 5 | Average |
|-------------------------------|-------------------|-------------------|-------------------|---------|
|                               | %RSD              | %RSD              | %RSD              | %RSD    |
| Cl <sup>-</sup>               | 0.32              | 0.37              | 1.37              | 0.69    |
| SO <sub>4</sub> <sup>2-</sup> | 0.57              | 0.18              | 1.40              | 0.72    |
| Br <sup>-</sup>               | 7.92              | 6.16              | 6.18              | 6.75    |
| F <sup>-</sup>                | 2.92              | 3.80              | 4.04              | 3.59    |



**CH4-4:** Terminology for textural description of mafic rocks in thin section.

**Grain size:**

CRYPTOCRYSTALLINE <0.1 mm  
MICROCRYSTALLINE = 0.1 – 0.2 mm  
FINE GRAINED = 0.2 – 1 mm  
MEDIUM GRAINED = 1 – 5 mm  
COARSE GRAINED = 5 – 30 mm  
PEGMATITIC = >30 mm

**Texture:**

GLOMEROPORPHYRITIC = large crystals that are surrounded by finer-grained matrix (phenocrysts) cluster and grow together.

GRANOBLASTIC = approximately equidimensional grains have sutured boundaries, and meet at ~120° triple junctions; characteristic of recrystallization.

INTERGRANULAR = angular interstices between feldspars occupied by pyroxene granules (very small grains).

INTERSERTAL = interstices filled with a mixture of glass and some pyroxene.

SUBOPHITIC = larger plagioclase are partially enclosed by pyroxene or olivine; the plagioclase laths typically impinge on one another to form sharp angles.

VARIOLITIC = divergent plagioclase laths with interstitial glass or intergrown with pyroxene granules

VESICULAR = cavities from expanded gases (vesicles) cover less than half of the specimen volume.

**CH4-5:** Inductively coupled plasma mass spectrometry (ICP-MS) methodology to analyse major, tracer, and rare earth elements. The method includes the preparation of silicate whole rocks via dissolution.

#### **Dissolution of silicate whole rocks- “mother solution”**

- 1) Accurately weigh the appropriate amount of powdered samples into an acid cleaned Savillex pot (usually 50 - 100 mg).
- 2) Add 10 drops of 15M HNO<sub>3</sub> (sb) per 10 mg of powdered sample to make a slurry.
- 3) In a scrubbed fume cupboard add ~1 drop of Artisar HF per mg of powdered sample.
- 4) Firmly screw on the lid and place on a hotplate set at 120 to 130 °C for at least 12 hours.
- 5) Remove samples from hotplate to cool.
- 6) Once cold, carefully remove lids (in a scrubbed fume cupboard) and check no sample remains undissolved and place back on the hotplate (lids off) and wait for samples to dry to incipient dryness (do not overdry).
- 7) If undissolved sample was present, repeat the HF - step 3.
- 8) Once dry and no sample is left undissolved, carefully remove from hotplate and add sufficient 6M HCl (sb) to dissolve the sample (at least 2 mL).
- 9) Replace the lid and leave on hotplate to reflux until all sample has dissolved.
- 10) If undissolved sample remains, add additional HCl and leave to reflux. If undissolved material still remains, dry down and carry out repeat attacks of concentrated HNO<sub>3</sub> (sb) and concentrated HCl (sb).
- 11) Once the sample is fully dissolved in 6M HCl (sb), remove lids and allow to dry to incipient dryness, be careful not to overdry.
- 12) Once dry add sufficient 6M HCl (sb) and leave to dissolve for at least 2 hours.
- 13) Transfer sample to a labelled and weighed acid cleaned HDPE bottle and rinse out the Savillex pot thoroughly with 6M HCl (sb) and Milli-Q water to make the solution (“mother solution”) up to an appropriate volume then reweigh.

Dilution factors of the *mother solutions* were calculated (usually on the order of 400).

At least one laboratory blank accompanied each batch of samples and this underwent the same digestion procedure as the samples, but without the addition of any sample powder.

#### **Dilution for tracer and rare earth elements analysis – “daughter solution-4’000”**

*Daughter solutions-4’000* for ICP-MS analysis were produced by subsampling the *mother solutions* (~1 mL) of dissolved rock, drying down on a hot plate and redissolving in 3 % (0.68 M) HNO<sub>3</sub> (sb) spiked with Indium and Rhenium (5 ppb) and Beryllium (20 ppb) internal standards, to give ~4000 times dilution for rocks.

#### **Dilution for major elements analysis – “daughter solution-80’000”**

*Daughter solutions-80’000* for ICP-MS analysis were produced by subsampling the *daughter solutions-4’000* (~250 µL), and redissolving in 3 % (0.68 M) HNO<sub>3</sub> (sb) spiked with Indium and Rhenium (5 ppb) and Beryllium (20 ppb) internal standards, to give ~80000 times dilution for rocks.

#### **ICP-MS analysis procedure**

Major, trace and rare earth elements (REE) were analysed on the Thermo X-Series 2 ICP-MS at the University of Southampton in solution mode. Before each analysis the instrument was tuned using a multi-element solution (Co, Y, In, La, Re, Bi and U) in order to obtain optimum sensitivity and stability. After tuning, the instrument was left to stabilise for at least a further 30 minutes. Samples were run in random order with six calibration

standards and a blank acid at the start and end of each run. At the very end of the run, several blank acids were analysed to constrain detection limits.

International standard JA2 was used to test for accuracy, and this standard as well as internal standards BAS206 and BRR1 were used to assess precision. Accuracy and precision of measurements are shown in Tables A.1 and A.2, respectively. Certified values for JA2 are from the online GeoREM database (<http://georoc.mpchmainz.gwdg.de>). Internal precision was monitored by measuring each sample four times and calculating the percentage relative standard deviation (%RSD) which is expressed as the standard deviation of the mean as a percentage of the mean. Data with RSD of >10 % were rejected. Results were measured as counts per second (CPS), these were processed using the Plasmalab software which calibrated counts to concentration using the calibration standards BHV2, BIR1, JB1a, JB3, and JGb1, . Further processing that was applied using this software package included blank, drift and matrix corrections. Internal drift was monitored using the spikes added to each solution (In, Re and Be). In runs where REEs were measured further data processing involved correction for REE oxide formation using results from Ba and REE single tuning solutions that allow oxide formation during each run to be measured and corrected for.

**Table A.1 - Accuracy JA2**

| Element |            | cert. values | average<br>N = 30-70 | average<br>N = 20-30 | average accuracy<br>% |
|---------|------------|--------------|----------------------|----------------------|-----------------------|
| Li      | µg/g = ppm | 29.1         | 29.0                 |                      | 4.31                  |
| Na      | µg/g       | 23100        |                      | 23900                | 7.26                  |
| Mg      | µg/g       | 45800        |                      | 44600                | 7.18                  |
| Al      | µg/g       | 81600        |                      | 84400                | 7.04                  |
| K       | µg/g       | 15000        |                      | 16900                | 9.20                  |
| Ca      | µg/g       | 45000        | 45600                |                      | 8.62                  |
| Sc      | µg/g       | 18.4         | 18.2                 |                      | 9.83                  |
| Ti      | µg/g       | 4169         | 4000                 |                      | 5.92                  |
| V       | µg/g       | 122          | 113                  |                      | 6.47                  |
| Cr      | µg/g       | 450          | 389                  |                      | 6.22                  |
| Mn      | µg/g       | 852          | 800                  |                      | 23.7                  |
| Fe      | µg/g       | 43400        |                      | 43000                | 10.9                  |
| Co      | µg/g       | 27.0         | 27.7                 |                      | 3.96                  |
| Ni      | µg/g       | 134          | 125                  |                      | 3.82                  |
| Cu      | µg/g       | 27.9         | 30.4                 |                      | 5.51                  |
| Zn      | µg/g       | 65.0         | 72.4                 |                      | 12.1                  |
| Rb      | µg/g       | 71.0         | 76.4                 |                      | 5.45                  |
| Sr      | µg/g       | 250          | 246                  |                      | 5.26                  |
| Y       | µg/g       | 18.1         | 17.5                 |                      | 5.38                  |
| Zr      | µg/g       | 112          | 120                  |                      | 7.63                  |
| Nb      | µg/g       | 9.00         | 9.00                 |                      | 4.50                  |
| Cs      | µg/g       | 4.90         | 5.18                 |                      | 6.01                  |
| Ba      | µg/g       | 315          | 323                  |                      | 6.44                  |
| La      | µg/g       | 16.1         | 16.1                 |                      | 5.07                  |
| Ce      | µg/g       | 33.7         | 33.9                 |                      | 5.34                  |
| Pr      | µg/g       | 3.70         | 3.84                 |                      | 3.76                  |
| Nd      | µg/g       | 14.2         | 14.6                 |                      | 3.54                  |
| Sm      | µg/g       | 3.10         | 3.12                 |                      | 3.71                  |
| Eu      | µg/g       | 0.91         | 0.91                 |                      | 3.36                  |
| Gd      | µg/g       | 3.00         | 3.06                 |                      | 3.74                  |
| Tb      | µg/g       | 0.48         | 0.48                 |                      | 4.21                  |
| Dy      | µg/g       | 2.90         | 2.93                 |                      | 4.44                  |
| Ho      | µg/g       | 0.61         | 0.60                 |                      | 4.49                  |
| Er      | µg/g       | 1.70         | 1.72                 |                      | 5.22                  |
| Tm      | µg/g       | 0.26         | 0.26                 |                      | 4.83                  |
| Yb      | µg/g       | 1.68         | 1.70                 |                      | 5.33                  |
| Lu      | µg/g       | 0.25         | 0.26                 |                      | 4.59                  |
| Hf      | µg/g       | 2.93         | 2.96                 |                      | 8.41                  |
| Ta      | µg/g       | 0.70         | 0.76                 |                      | 19.2                  |
| Pb      | µg/g       | 19.3         | 22.0                 |                      | 10.4                  |
| Th      | µg/g       | 5.00         | 4.90                 |                      | 5.55                  |
| U       | µg/g       | 2.20         | 2.26                 |                      | 4.66                  |

**Table A.2 - Precision**

| Element | JA2<br>%RSD | BAS206<br>%RSD | BRR-1<br>%RSD | Average<br>%RSD |
|---------|-------------|----------------|---------------|-----------------|
| Li      | 2.16        | 5.14           | 4.95          | 4.08            |
| Na      | 3.10        | 2.43           | 2.02          | 2.52            |
| Mg      | 3.79        | 2.63           | 2.11          | 2.84            |
| Al      | 3.49        | 2.46           | 2.78          | 2.91            |
| K       | 3.92        | 5.02           | 16.4          | 8.44            |
| Ca      | 4.28        | 2.13           | 2.68          | 3.03            |
| Sc      | 5.10        | 2.63           | 2.18          | 3.30            |
| Ti      | 3.11        | 1.75           | 2.66          | 2.50            |
| V       | 3.52        | 2.88           | 3.23          | 3.21            |
| Cr      | 3.63        | 6.07           | 3.43          | 4.38            |
| Mn      | 13.1        | 2.22           | 2.33          | 5.89            |
| Fe      | 5.62        | 2.77           | 2.76          | 3.72            |
| Co      | 1.97        | 1.49           | 1.85          | 1.77            |
| Ni      | 2.09        | 2.60           | 2.21          | 2.30            |
| Cu      | 2.56        | 2.14           | 6.37          | 3.69            |
| Zn      | 78.1        | 8.91           | 8.19          | 31.7            |
| Rb      | 2.61        | 11.0           | 20.9          | 11.5            |
| Sr      | 2.73        | 7.50           | 9.25          | 6.49            |
| Y       | 2.85        | 2.66           | 2.34          | 2.62            |
| Zr      | 3.51        | 2.89           | 6.40          | 4.27            |
| Nb      | 2.30        | 614            | 4.02          | 207             |
| Cs      | 2.93        | 4.84           | 15.0          | 7.60            |
| Ba      | 3.25        | 11.6           | 7.70          | 7.53            |
| La      | 2.62        | 5.94           | 1.74          | 3.43            |
| Ce      | 2.75        | 5.79           | 1.90          | 3.48            |
| Pr      | 1.87        | 5.73           | 1.27          | 2.96            |
| Nd      | 1.77        | 5.68           | 1.12          | 2.86            |
| Sm      | 1.88        | 5.54           | 1.32          | 2.91            |
| Eu      | 1.71        | 5.77           | 1.19          | 2.89            |
| Gd      | 1.83        | 6.66           | 1.29          | 3.26            |
| Tb      | 2.26        | 5.93           | 1.61          | 3.26            |
| Dy      | 2.26        | 5.67           | 1.28          | 3.07            |
| Ho      | 2.30        | 5.59           | 1.36          | 3.08            |
| Er      | 2.61        | 5.77           | 1.67          | 3.35            |
| Tm      | 2.92        | 5.80           | 1.80          | 3.51            |
| Yb      | 2.68        | 5.80           | 1.78          | 3.42            |
| Lu      | 2.71        | 5.90           | 1.77          | 3.46            |
| Hf      | 3.53        | 19.5           | 18.0          | 13.7            |
| Ta      | 8.99        | 11.0           | 219           | 79.8            |
| Pb      | 4.75        | 14.7           | 56.3          | 25.3            |
| Th      | 2.94        | 10.2           | 85.8          | 33.0            |
| U       | 2.34        | 6.30           | 3.88          | 4.17            |

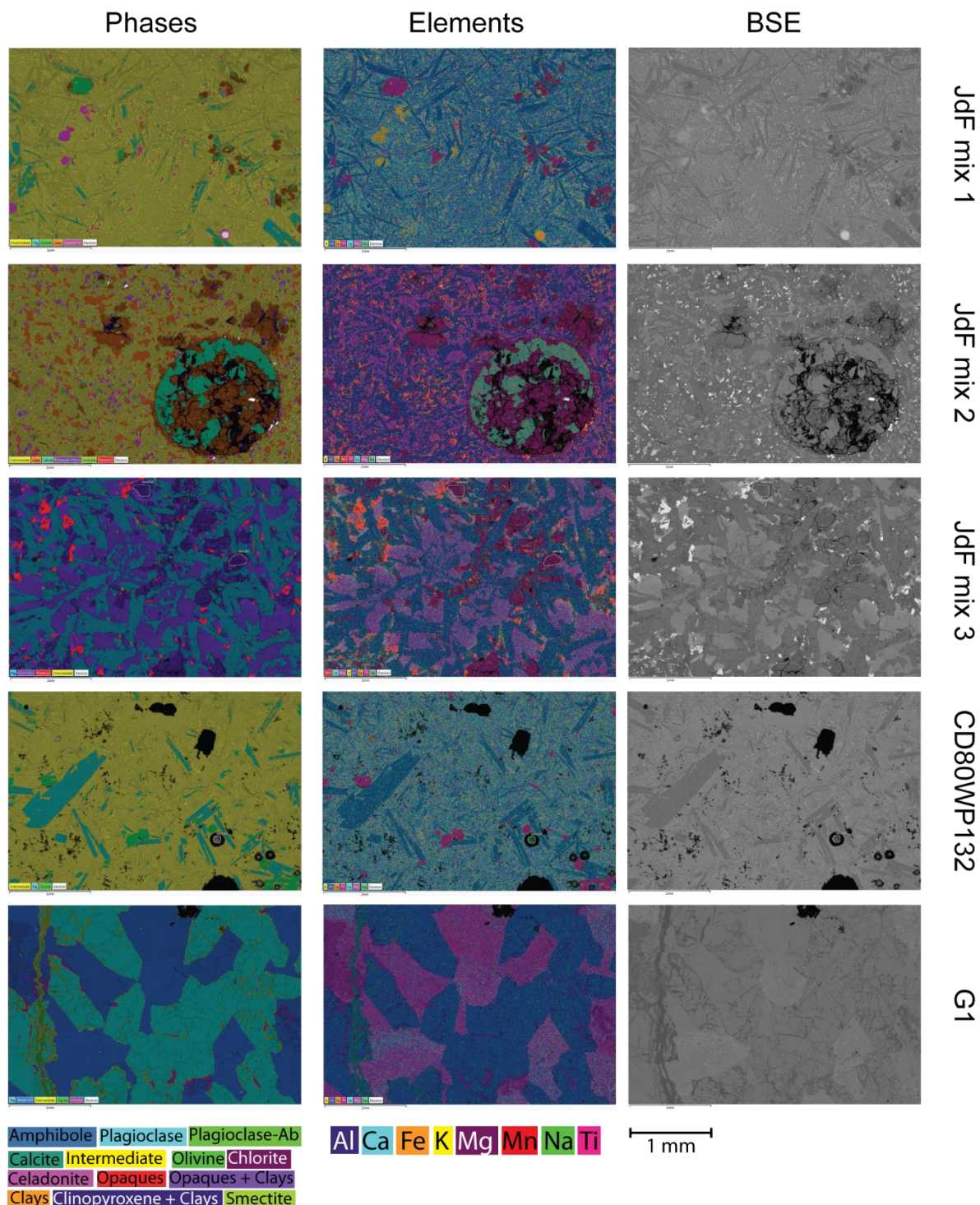
## A-Chapter 4, Analyses

**CH4-6:** Summary of SEM analyses on whole rocks.

SEM images are shown divided for each rock in BSE-back scatter electron, elemental line, and mineral phase mode.

SEM-EDS analyses and correspondent analytical error (standard deviation,  $\sigma$ ) are shown for both whole rock and single phases. To note: the total for SEM-EDS analyses is forced to be 100 % from the software.

Accuracy and precision for SEM-EDS analyses have been calibrated with the standard glass sample BIR1G.



## DATA FOR OXIDES ANALYSES SEM-EDS (%)

|                          | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | MnO  | MgO   | CaO   | Na <sub>2</sub> O | K <sub>2</sub> O | P <sub>2</sub> O <sub>5</sub> | SO <sub>3</sub> | CO <sub>2</sub> |
|--------------------------|------------------|------------------|--------------------------------|--------------------------------|------|-------|-------|-------------------|------------------|-------------------------------|-----------------|-----------------|
| <b>JdF mix 1</b>         |                  |                  |                                |                                |      |       |       |                   |                  |                               |                 |                 |
| Bulk rock                | 49.88            | 1.45             | 16.07                          | 11.20                          | 0.14 | 6.88  | 10.70 | 3.18              | 0.49             |                               |                 |                 |
| Intermediate             | 50.23            | 1.52             | 15.94                          | 11.05                          |      | 6.44  | 11.11 | 3.26              | 0.45             |                               |                 |                 |
| Plagioclase-labradorite  | 50.16            |                  | 31.44                          | 1.11                           |      | 0.41  | 12.98 | 3.90              |                  |                               |                 |                 |
| Olivine                  | 40.37            |                  |                                | 16.66                          |      | 42.53 | 0.45  |                   |                  |                               |                 |                 |
| Clays                    | 53.12            |                  | 2.85                           | 17.54                          |      | 22.63 | 1.66  | 1.19              | 1.01             |                               |                 |                 |
| Celadonite               | 44.95            | 2.90             | 4.03                           | 37.23                          |      | 4.02  | 1.50  | 1.56              | 3.82             |                               |                 |                 |
| <b>JdF mix 2</b>         |                  |                  |                                |                                |      |       |       |                   |                  |                               |                 |                 |
| Bulk rock                | 48.94            | 1.39             | 14.86                          | 9.98                           | 0.36 | 8.55  | 12.08 | 3.26              | 0.20             |                               | 0.38            |                 |
| Intermediate             | 52.91            | 0.78             | 20.97                          | 6.06                           |      | 4.81  | 9.63  | 4.54              | 0.17             |                               | 0.13            |                 |
| Clays                    | 52.79            | 0.86             | 3.90                           | 14.33                          | 0.16 | 18.79 | 8.40  | 0.55              | 0.22             |                               |                 |                 |
| Smectite                 | 79.75            |                  | 5.26                           | 5.83                           |      | 4.80  | 1.62  | 2.31              | 0.43             |                               |                 |                 |
| Opakes                   | 8.62             | 20.38            | 3.30                           | 64.39                          |      | 2.05  |       | 1.25              |                  |                               |                 |                 |
| Oxides + clays           | 33.24            | 10.18            | 6.83                           | 35.14                          | 0.39 | 6.84  | 3.34  | 2.75              | 0.37             | 0.32                          | 0.60            |                 |
| Calcite                  | 0.61             |                  |                                | 0.86                           | 2.74 | 0.83  | 36.78 |                   |                  |                               | 0.59            | 57.54           |
| <b>JdF mix 3</b>         |                  |                  |                                |                                |      |       |       |                   |                  |                               |                 |                 |
| Bulk rock                | 50.58            | 1.16             | 17.21                          | 8.74                           | 0.11 | 7.70  | 10.57 | 3.57              | 0.23             | 0.13                          |                 |                 |
| Intermediate             | 44.69            |                  | 15.69                          | 15.79                          |      | 6.30  | 13.95 | 3.58              |                  |                               |                 |                 |
| Plagioclase-labradorite1 | 53.18            |                  | 27.17                          | 2.25                           |      | 1.42  | 10.53 | 5.31              | 0.15             |                               |                 |                 |
| Plagioclase-labradorite2 | 52.06            |                  | 31.05                          |                                |      |       | 12.29 | 4.60              |                  |                               |                 |                 |
| Clinopyroxene            | 51.12            | 0.86             | 2.78                           | 12.08                          |      | 15.80 | 17.36 |                   |                  |                               |                 |                 |
| Clinopyroxene + clays    | 52.28            |                  | 5.34                           | 13.52                          |      | 15.58 | 11.82 | 1.47              |                  |                               |                 |                 |
| Vesicle 1 - saponite     | 55.42            |                  | 4.95                           | 15.45                          |      | 21.10 | 0.85  | 1.72              | 0.50             |                               |                 |                 |
| Vesicle 2 - saponite     | 54.48            |                  | 5.94                           | 16.72                          |      | 21.45 | 1.41  |                   |                  |                               |                 |                 |
| Opakes                   | 6.25             | 24.42            | 2.84                           | 63.72                          |      | 1.76  | 0.80  |                   | 0.20             |                               |                 |                 |
| <b>CD80WP132</b>         |                  |                  |                                |                                |      |       |       |                   |                  |                               |                 |                 |
| Bulk rock                | 49.43            | 1.13             | 16.78                          | 11.45                          | 0.15 | 6.76  | 11.27 | 2.71              | 0.07             |                               | 0.24            |                 |
| Intermediate             | 50.1             | 1.28             | 15.24                          | 12.48                          |      | 6.8   | 11.32 | 2.72              | 0.07             |                               |                 |                 |
| Plagioclase-labradorite  | 49.24            |                  | 31.96                          |                                |      | 0.44  | 14.31 | 3.18              |                  |                               |                 |                 |
| Vesicle 1                | 74.89            |                  |                                | 0.87                           |      | 4.10  | 9.06  | 11.95             |                  |                               |                 |                 |
| Olivine                  | 38.48            |                  | 0.78                           | 20.24                          |      | 39.73 | 0.78  |                   |                  |                               |                 |                 |
| <b>G1</b>                |                  |                  |                                |                                |      |       |       |                   |                  |                               |                 |                 |
| Bulk rock                | 48.3             | 0.09             | 23.31                          | 5.26                           |      | 9.67  | 12.76 | 1.46              | 0.15             |                               |                 |                 |
| Intermediate             | 49.83            |                  | 26.85                          | 2.78                           |      | 8.27  | 5.21  | 6.53              | 0.53             |                               |                 |                 |
| Plagioclase-anorthite    | 44.54            |                  | 35.84                          | 0.68                           |      | 0.93  | 16.6  | 1.26              | 0.16             |                               |                 |                 |
| Plagioclase-albite       | 57.00            |                  | 26.34                          |                                |      |       | 0.60  | 15.81             | 0.25             |                               |                 |                 |
| Amphibole-actinolite     | 54.29            | 0.22             | 4.53                           | 8.72                           | 0.17 | 20.89 | 10.57 | 0.52              | 0.09             |                               |                 |                 |
| Chlorite                 | 39.75            |                  | 19.19                          | 9.67                           |      | 29.50 | 1.42  | 0.48              |                  |                               |                 |                 |

## ERROR (σ) FOR OXIDES ANALYSES SEM-EDS (%)

|                          | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | MnO  | MgO  | CaO  | Na <sub>2</sub> O | K <sub>2</sub> O | P <sub>2</sub> O <sub>5</sub> | SO <sub>3</sub> | CO <sub>2</sub> |
|--------------------------|------------------|------------------|--------------------------------|--------------------------------|------|------|------|-------------------|------------------|-------------------------------|-----------------|-----------------|
| <b>JdF mix 1</b>         |                  |                  |                                |                                |      |      |      |                   |                  |                               |                 |                 |
| Bulk rock                | 0.04             | 0.01             | 0.03                           | 0.03                           | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               |                 |                 |
| Intermediate             | 0.04             | 0.02             | 0.03                           | 0.03                           |      | 0.02 | 0.02 | 0.02              | 0.01             |                               |                 |                 |
| Plagioclase-labradorite  | 0.22             |                  | 0.16                           | 0.08                           |      | 0.06 | 0.10 | 0.09              |                  |                               |                 |                 |
| Olivine                  | 0.38             |                  |                                | 0.30                           | 0.33 |      | 0.08 |                   |                  |                               |                 |                 |
| Clays                    | 0.33             |                  | 0.14                           | 0.24                           |      | 0.22 | 0.08 | 0.11              | 0.06             |                               |                 |                 |
| Celadonite               | 0.38             | 0.15             | 0.17                           | 0.36                           |      | 0.17 | 0.10 | 0.16              | 0.10             |                               |                 |                 |
| <b>JdF mix 2</b>         |                  |                  |                                |                                |      |      |      |                   |                  |                               |                 |                 |
| Bulk rock                | 0.05             | 0.02             | 0.03                           | 0.03                           | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.01            |                 |
| Intermediate             | 0.06             | 0.02             | 0.04                           | 0.03                           |      | 0.02 | 0.02 | 0.03              | 0.01             |                               | 0.02            |                 |
| Clays                    | 0.09             | 0.03             | 0.04                           | 0.06                           | 0.02 | 0.06 | 0.03 | 0.03              | 0.01             |                               |                 |                 |
| Smectite                 | 0.60             |                  | 0.24                           | 0.29                           |      | 0.22 | 0.13 | 0.20              | 0.09             |                               |                 |                 |
| Opakes                   | 0.44             | 0.55             | 0.37                           | 0.85                           |      | 0.34 |      | 0.38              |                  |                               |                 |                 |
| Oxides + clays           | 0.18             | 0.11             | 0.11                           | 0.19                           | 0.06 | 0.11 | 0.06 | 0.10              | 0.03             | 0.08                          | 0.07            |                 |
| Calcite                  | 0.04             |                  |                                | 0.06                           | 0.06 | 0.04 | 0.18 |                   |                  |                               | 0.04            | 0.71            |
| <b>JdF mix 3</b>         |                  |                  |                                |                                |      |      |      |                   |                  |                               |                 |                 |
| Bulk rock                | 0.04             | 0.01             | 0.03                           | 0.03                           | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             | 0.02                          |                 |                 |
| Intermediate             | 0.89             |                  | 0.59                           | 0.74                           |      | 0.44 | 0.44 | 0.40              |                  |                               |                 |                 |
| Plagioclase-labradorite1 | 0.06             |                  | 0.04                           | 0.03                           |      | 0.02 | 0.03 | 0.03              | 0.01             |                               |                 |                 |
| Plagioclase-labradorite2 | 1.00             |                  | 0.73                           |                                |      |      | 0.44 | 0.41              |                  |                               |                 |                 |
| Clinopyroxene            | 0.50             | 0.16             | 0.21                           | 0.35                           |      | 0.31 | 0.26 |                   |                  |                               |                 |                 |
| Clinopyroxene + clays    | 0.07             |                  | 0.03                           | 0.05                           |      | 0.04 | 0.03 | 0.02              |                  |                               |                 |                 |
| Vesicle 1 - saponite     | 0.99             |                  | 0.48                           | 0.70                           |      | 0.64 | 0.21 | 0.36              | 0.16             |                               |                 |                 |
| Vesicle 2 - saponite     | 1.04             |                  | 0.50                           | 0.74                           |      | 0.67 | 0.23 |                   |                  |                               |                 |                 |
| Opakes                   | 0.17             | 0.26             | 0.15                           | 0.36                           |      | 0.15 | 0.07 |                   | 0.05             |                               |                 |                 |
| <b>CD80WP132</b>         |                  |                  |                                |                                |      |      |      |                   |                  |                               |                 |                 |
| Bulk rock                | 0.04             | 0.01             | 0.03                           | 0.03                           | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.01            |                 |
| Intermediate             | 0.05             | 0.02             | 0.03                           | 0.03                           |      | 0.02 | 0.02 | 0.02              | 0.01             |                               |                 |                 |
| Plagioclase-labradorite  | 0.13             |                  | 0.10                           | 0.05                           |      | 0.04 | 0.06 | 0.05              |                  |                               |                 |                 |
| Vesicle 1                | 1.86             |                  |                                |                                |      | 0.72 | 0.70 | 1.01              |                  |                               |                 |                 |
| Olivine                  | 0.28             |                  | 0.11                           | 0.24                           |      | 0.25 | 0.06 |                   |                  |                               |                 |                 |
| <b>G1</b>                |                  |                  |                                |                                |      |      |      |                   |                  |                               |                 |                 |
| Bulk rock                | 0.04             | 0.01             | 0.03                           | 0.02                           |      | 0.02 | 0.02 | 0.01              | 0.01             |                               |                 |                 |
| Intermediate             | 0.18             |                  | 0.13                           | 0.08                           |      | 0.09 | 0.06 | 0.08              | 0.03             |                               |                 |                 |
| Plagioclase-anorthite    | 0.06             |                  | 0.05                           | 0.02                           |      | 0.02 | 0.03 | 0.02              | 0.01             |                               |                 |                 |
| Plagioclase-albite       | 0.43             |                  | 0.30                           |                                |      |      | 0.08 | 0.24              | 0.06             |                               |                 |                 |
| Amphibole-actinolite     | 0.07             | 0.02             | 0.03                           | 0.04                           | 0.02 | 0.04 | 0.03 | 0.02              | 0.01             |                               |                 |                 |
| Chlorite                 | 0.35             |                  | 0.27                           | 0.23                           |      | 0.27 | 0.08 | 0.11              |                  |                               |                 |                 |

STANDARD BIR1G ANALYSES:

|                                    | <b>Certified value</b><br><b>(%)</b> | <b>Average N=13</b><br><b>(%)</b> | <b>Accuracy</b><br><b>(%)</b> | <b>Precision</b><br><b>% RSD</b> |
|------------------------------------|--------------------------------------|-----------------------------------|-------------------------------|----------------------------------|
| <b>SiO<sub>2</sub></b>             | 47.3                                 | 47.2                              | 1.06                          | 1.14                             |
| <b>TiO<sub>2</sub></b>             | 0.98                                 | 1.04                              | 9.70                          | 7.59                             |
| <b>Al<sub>2</sub>O<sub>3</sub></b> | 15.6                                 | 15.6                              | 1.37                          | 1.30                             |
| <b>Fe<sub>2</sub>O<sub>3</sub></b> | 10.5                                 | 11.6                              | 2.10                          | 1.36                             |
| <b>MnO</b>                         | 0.18                                 | 0.23                              | 7.86                          | 4.04                             |
| <b>MgO</b>                         | 9.37                                 | 9.75                              | 1.72                          | 1.37                             |
| <b>CaO</b>                         | 13.4                                 | 13.3                              | 1.86                          | 1.32                             |
| <b>Na<sub>2</sub>O</b>             | 1.89                                 | 1.90                              | 4.34                          | 2.91                             |
| <b>K<sub>2</sub>O</b>              | 0.03                                 | -                                 | -                             | -                                |



**CH4-7:** Inductively coupled plasma mass spectrometry (ICP-MS) analyses on major, tracer, and rare earth elements, for both grain size fractions (63-125  $\mu\text{m}$  and <63  $\mu\text{m}$ ). The element composition is given in  $\mu\text{g/g}$ . Analytical errors are reported in a separated table for both grain size fractions, and are given in percentage relative standard deviation (%RSD), which is expressed as the standard deviation of the mean as a percentage of the mean.

DATA (ppm)

| ID        | JdF mix 1            |                   | JdF mix 2            |                   | JdF mix 3            |                   | CD80WP132            |                   | G1                   |                   |        |
|-----------|----------------------|-------------------|----------------------|-------------------|----------------------|-------------------|----------------------|-------------------|----------------------|-------------------|--------|
|           | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | 63-125 $\mu\text{m}$ | <63 $\mu\text{m}$ | CLEAN  |
| <b>Na</b> | 21200                | 24500             | 22200                | 24700             | 22600                | 23600             | 18000                | 17200             | 5430                 | 3680              | 4910   |
| <b>Mg</b> | 44100                | 51000             | 52200                | 54500             | 50900                | 48000             | 56900                | 54800             | 75200                | 70500             | 80900  |
| <b>Al</b> | 80300                | 95200             | 90300                | 101000            | 89300                | 92600             | 83400                | 84200             | 114000               | 69000             | 102000 |
| <b>K</b>  | 1820                 | 2160              | 1240                 | 1360              | 1880                 | 1640              | 337                  | 267               | 322                  | 196               | 269    |
| <b>Ca</b> | 87800                | 104000            | 102000               | 113000            | 104000               | 103000            | 101000               | 98300             | 129000               | 83400             | 121000 |
| <b>Mn</b> | 1930                 | 2180              | 1710                 | 1820              | 1880                 | 1880              | 1970                 | 1770              | 916                  | 836               | 960    |
| <b>Fe</b> | 78100                | 80000             | 77600                | 81900             | 83400                | 82900             | 114000               | 85300             | 37500                | 34400             | 40100  |
| <b>Sr</b> | 119                  | 141               | 127                  | 141               | 125                  | 135               | 75.4                 | 73.8              | 56.3                 | 37.5              | 49.5   |
| <b>Li</b> | 6.02                 | 5.59              | 5.37                 | 5.27              | 5.43                 | 5.37              | 5.04                 | 4.67              | 5.39                 | 7.17              | 5.24   |
| <b>Sc</b> | 50.1                 | 47.8              | 51.3                 | 50.3              | 52.9                 | 49.7              | 53.1                 | 52.2              | 36.2                 | 40.3              | 39.6   |
| <b>Rb</b> | 3.56                 | 3.61              | 2.21                 | 2.23              | 4.43                 | 3.19              | 0.80                 | 0.73              | 1.00                 | 1.18              | 1.04   |
| <b>Sr</b> | 120                  | 118               | 114                  | 118               | 113                  | 117               | 67.1                 | 68.4              | 49.8                 | 45.0              | 45.2   |
| <b>Y</b>  | 48.0                 | 43.9              | 32.8                 | 30.8              | 32.8                 | 29.4              | 34.4                 | 31.6              | 2.79                 | 3.01              | 3.13   |
| <b>Zr</b> | 127                  | 114               | 78.9                 | 72.4              | 77.0                 | 68.4              | 61.9                 | 56.0              | 1.26                 | 1.39              | 1.47   |
| <b>Nb</b> | 3.12                 | 2.91              | 2.05                 | 1.88              | 2.16                 | 2.12              | 1.50                 | 1.37              | 0.14                 | 0.14              | 0.15   |
| <b>Sn</b> | 1.28                 | 1.19              | 1.01                 | 0.92              | 1.05                 | 0.98              | 0.81                 | 0.85              | 0.23                 | 0.23              | 0.25   |
| <b>Cs</b> | 0.14                 | 0.14              | 0.08                 | 0.08              | 0.13                 | 0.10              | 0.01                 | 0.01              | 0.03                 | 0.03              | 0.03   |
| <b>Ba</b> | 17.0                 | 14.3              | 8.63                 | 8.35              | 8.98                 | 7.97              | 9.36                 | 8.62              | 2.91                 | 3.01              | 3.24   |
| <b>La</b> | 4.36                 | 3.91              | 2.53                 | 2.25              | 2.36                 | 2.00              | 1.78                 | 1.61              | 0.06                 | 0.06              | 0.07   |
| <b>Ce</b> | 13.7                 | 12.3              | 8.15                 | 7.29              | 7.65                 | 6.51              | 5.90                 | 5.33              | 0.15                 | 0.17              | 0.18   |
| <b>Pr</b> | 2.43                 | 2.19              | 1.48                 | 1.35              | 1.40                 | 1.21              | 1.12                 | 1.02              | 0.03                 | 0.03              | 0.03   |
| <b>Nd</b> | 13.3                 | 12.0              | 8.29                 | 7.66              | 7.99                 | 6.94              | 6.70                 | 6.07              | 0.20                 | 0.20              | 0.22   |
| <b>Sm</b> | 4.70                 | 4.29              | 3.11                 | 2.88              | 3.05                 | 2.70              | 2.71                 | 2.49              | 0.12                 | 0.13              | 0.14   |
| <b>Eu</b> | 1.64                 | 1.52              | 1.19                 | 1.16              | 1.21                 | 1.13              | 1.04                 | 0.96              | 0.08                 | 0.08              | 0.09   |
| <b>Gd</b> | 6.57                 | 6.03              | 4.46                 | 4.17              | 4.41                 | 3.95              | 4.15                 | 3.83              | 0.26                 | 0.27              | 0.30   |
| <b>Tb</b> | 1.20                 | 1.10              | 0.82                 | 0.78              | 0.83                 | 0.74              | 0.80                 | 0.74              | 0.06                 | 0.06              | 0.06   |
| <b>Dy</b> | 7.96                 | 7.28              | 5.49                 | 5.17              | 5.56                 | 4.98              | 5.57                 | 5.11              | 0.45                 | 0.47              | 0.50   |
| <b>Ho</b> | 1.71                 | 1.57              | 1.20                 | 1.13              | 1.20                 | 1.09              | 1.24                 | 1.14              | 0.10                 | 0.11              | 0.12   |
| <b>Er</b> | 5.00                 | 4.61              | 3.50                 | 3.29              | 3.52                 | 3.17              | 3.74                 | 3.44              | 0.33                 | 0.35              | 0.37   |
| <b>Tm</b> | 0.76                 | 0.70              | 0.53                 | 0.50              | 0.53                 | 0.48              | 0.58                 | 0.53              | 0.05                 | 0.06              | 0.06   |
| <b>Yb</b> | 4.91                 | 4.50              | 3.44                 | 3.25              | 3.48                 | 3.15              | 3.81                 | 3.50              | 0.35                 | 0.39              | 0.39   |
| <b>Lu</b> | 0.74                 | 0.68              | 0.53                 | 0.50              | 0.53                 | 0.48              | 0.58                 | 0.54              | 0.05                 | 0.06              | 0.06   |
| <b>Hf</b> | 3.49                 | 3.17              | 2.32                 | 2.16              | 2.26                 | 2.04              | 1.86                 | 1.72              | 0.06                 | 0.07              | 0.07   |
| <b>Ta</b> | 0.21                 | 0.20              | 0.14                 | 0.13              | 0.14                 | 0.14              | 0.10                 | 0.09              | 0.01                 | 0.01              | 0.01   |
| <b>Pb</b> | 0.65                 | 0.66              | 0.42                 | 0.38              | 0.44                 | 0.39              | 0.30                 | 0.37              | 0.15                 | 0.17              | 0.17   |
| <b>Th</b> | 0.21                 | 0.19              | 0.12                 | 0.11              | 0.11                 | 0.09              | 0.10                 | 0.09              | 0.01                 | 0.01              | 0.01   |
| <b>U</b>  | 0.09                 | 0.08              | 0.07                 | 0.06              | 0.08                 | 0.07              | 0.05                 | 0.06              | 0.01                 | 0.01              | 0.01   |

# ERROR (% RSD)

| ID | JdF mix 1 - %RSD |       | JdF mix 2 - %RSD |       | JdF mix 3 - %RSD |       | CD80WP132 - %RSD |       | G1 - %RSD    |       |                |
|----|------------------|-------|------------------|-------|------------------|-------|------------------|-------|--------------|-------|----------------|
|    | 63-125<br>µm     | <63µm | 63-125<br>µm     | <63µm | 63-125<br>µm     | <63µm | 63-125<br>µm     | <63µm | 63-125<br>µm | <63µm | <63µm<br>CLEAN |
| Na | 0.54             | 0.74  | 1.02             | 0.45  | 0.59             | 0.39  | 0.27             | 0.55  | 0.95         | 0.37  | 1.31           |
| Mg | 0.54             | 0.56  | 0.30             | 0.32  | 0.39             | 0.39  | 0.29             | 0.60  | 0.34         | 0.27  | 0.51           |
| Al | 0.37             | 0.20  | 0.46             | 0.44  | 0.78             | 0.48  | 0.72             | 0.39  | 0.44         | 0.66  | 0.47           |
| K  | 1.51             | 1.04  | 2.61             | 3.57  | 1.64             | 1.41  | 3.06             | 17.65 | 6.32         | 10.06 | 11.13          |
| Ca | 1.17             | 1.02  | 1.00             | 0.81  | 0.37             | 0.67  | 1.01             | 1.47  | 0.68         | 0.73  | 0.20           |
| Mn | 0.53             | 0.43  | 0.44             | 0.60  | 0.73             | 0.42  | 0.72             | 0.71  | 0.32         | 0.71  | 0.73           |
| Fe | 0.86             | 3.15  | 0.45             | 1.20  | 1.54             | 3.51  | 0.43             | 22.25 | 0.56         | 0.57  | 0.24           |
| Sr | 1.19             | 1.23  | 1.14             | 1.55  | 1.80             | 0.76  | 1.13             | 2.87  | 2.09         | 3.01  | 2.30           |
|    |                  |       |                  |       |                  |       |                  |       |              |       |                |
| Li | 1.98             | 1.38  | 1.25             | 2.08  | 1.95             | 1.69  | 1.28             | 1.23  | 1.24         | 0.64  | 1.69           |
| Sc | 0.30             | 0.85  | 0.76             | 0.71  | 0.85             | 0.42  | 0.27             | 0.36  | 0.60         | 0.22  | 0.38           |
| Rb | 0.79             | 0.74  | 1.22             | 1.16  | 0.92             | 1.09  | 1.16             | 1.60  | 1.96         | 1.80  | 1.16           |
| Sr | 0.42             | 0.46  | 0.38             | 0.40  | 0.70             | 0.64  | 0.24             | 0.28  | 0.78         | 0.71  | 0.45           |
| Y  | 0.45             | 0.22  | 0.28             | 0.17  | 0.10             | 0.31  | 0.24             | 0.25  | 1.61         | 0.62  | 0.71           |
| Zr | 0.34             | 0.68  | 0.32             | 0.36  | 0.22             | 0.28  | 0.36             | 0.29  | 1.34         | 1.40  | 1.50           |
| Nb | 0.49             | 0.74  | 1.54             | 0.76  | 0.71             | 0.56  | 0.88             | 1.21  | 2.37         | 2.50  | 1.75           |
| Sn | 2.11             | 1.11  | 0.54             | 1.25  | 1.28             | 0.70  | 2.39             | 1.92  | 1.51         | 1.26  | 1.55           |
| Cs | 1.85             | 1.96  | 3.29             | 3.49  | 2.06             | 3.63  | 9.19             | 10.34 | 4.82         | 3.56  | 6.70           |
| Ba | 0.79             | 0.66  | 1.20             | 0.68  | 0.84             | 1.42  | 0.70             | 1.45  | 1.53         | 1.89  | 0.79           |
| La | 0.51             | 0.67  | 0.58             | 0.70  | 0.45             | 0.58  | 0.82             | 0.32  | 5.10         | 1.61  | 1.75           |
| Ce | 0.41             | 0.12  | 0.16             | 0.56  | 0.48             | 0.38  | 0.32             | 0.12  | 3.11         | 1.44  | 1.49           |
| Pr | 0.91             | 0.28  | 0.41             | 0.38  | 1.02             | 0.60  | 0.16             | 0.55  | 3.92         | 1.60  | 6.70           |
| Nd | 0.41             | 0.66  | 0.72             | 0.70  | 0.66             | 0.52  | 0.37             | 0.33  | 2.83         | 5.61  | 2.99           |
| Sm | 1.43             | 1.04  | 1.43             | 0.75  | 1.15             | 0.63  | 0.99             | 0.94  | 2.98         | 7.36  | 4.20           |
| Eu | 0.71             | 0.65  | 0.88             | 0.41  | 0.96             | 0.95  | 0.59             | 1.33  | 1.95         | 0.96  | 3.04           |
| Gd | 0.66             | 0.43  | 0.74             | 0.56  | 0.43             | 1.06  | 0.73             | 0.71  | 2.94         | 1.96  | 2.59           |
| Tb | 0.48             | 0.46  | 0.26             | 0.75  | 0.34             | 0.40  | 0.86             | 1.32  | 2.91         | 2.59  | 2.28           |
| Dy | 0.40             | 0.23  | 0.36             | 0.74  | 0.39             | 0.98  | 0.35             | 0.77  | 2.06         | 0.48  | 0.71           |
| Ho | 0.25             | 0.39  | 0.44             | 0.63  | 0.30             | 0.43  | 0.49             | 0.88  | 0.89         | 2.64  | 2.59           |
| Er | 0.26             | 0.44  | 0.27             | 0.73  | 0.37             | 0.95  | 1.02             | 0.49  | 2.18         | 1.87  | 1.18           |
| Tm | 0.49             | 0.75  | 0.63             | 1.15  | 0.19             | 0.84  | 0.97             | 0.82  | 3.66         | 3.06  | 0.79           |
| Yb | 0.22             | 0.36  | 0.38             | 0.16  | 0.90             | 0.80  | 0.28             | 0.39  | 1.83         | 1.05  | 1.34           |
| Lu | 0.20             | 0.31  | 0.62             | 0.66  | 0.38             | 0.95  | 0.29             | 0.31  | 4.90         | 2.92  | 1.59           |
| Hf | 0.38             | 0.27  | 0.59             | 0.46  | 0.59             | 0.80  | 0.40             | 0.80  | 2.88         | 2.66  | 1.44           |
| Ta | 1.19             | 1.40  | 0.82             | 1.72  | 2.51             | 0.97  | 1.67             | 1.34  | 5.80         | 1.92  | 10.07          |
| Pb | 0.99             | 1.16  | 1.21             | 1.47  | 1.12             | 2.12  | 0.87             | 1.91  | 0.62         | 2.90  | 1.87           |
| Th | 1.20             | 1.04  | 0.83             | 1.09  | 1.32             | 1.50  | 1.34             | 1.11  | 2.97         | 8.62  | 4.28           |
| U  | 2.76             | 2.00  | 2.58             | 2.39  | 4.12             | 2.10  | 3.10             | 1.52  | 1.82         | 5.41  | 4.81           |

**CH4-8:** X-ray diffraction (XRD) analyses on mineralogical composition of rocks, for both grain size fractions (63-125 µm and <63 µm).

MINERALOGICAL COMPOSITIONS (%)

| ID                                |           | Plagioclase -<br>anorthite | Plagioclase -<br>labradorite | Plagioclase -<br>andesine | Clinopyroxene -<br>augite | Clinopyroxene -<br>diopside | Amphibole | Chlorite | Talc | TOTAL |
|-----------------------------------|-----------|----------------------------|------------------------------|---------------------------|---------------------------|-----------------------------|-----------|----------|------|-------|
| <b>JdF mix 1</b>                  | 63-125 µm | 19.4                       | 15.7                         | 20.0                      | 42.2                      |                             |           |          |      | 97.3  |
|                                   | <63 µm    | 20.8                       | 15.2                         | 21.4                      | 43.7                      |                             |           |          |      | 101.1 |
| <b>JdF mix 2</b>                  | 63-125 µm | 21.9                       | 20.3                         | 19.7                      | 41.4                      |                             |           |          |      | 103.3 |
|                                   | <63 µm    | 21.1                       | 19.4                         | 22.4                      | 41.5                      |                             |           |          |      | 104.4 |
| <b>JdF mix 3</b>                  | 63-125 µm | 19.0                       | 27.5                         | 11.1                      | 43.0                      |                             |           |          |      | 100.6 |
|                                   | <63 µm    | 24.9                       | 32.3                         | 11.2                      | 41.2                      |                             |           |          |      | 109.6 |
| <b>CD80WP132<br/>(normalized)</b> | 63-125 µm | 55.5                       |                              |                           |                           | 44.5                        |           |          |      | 100   |
|                                   | <63 µm    | 57.8                       |                              |                           |                           | 42.2                        |           |          |      | 100   |
| <b>G1</b>                         | 63-125 µm | 76.0                       |                              |                           |                           | 12.8                        | 23.0      | 1.0      | 4.3  | 117.1 |
|                                   | <63 µm    | 58.8                       |                              |                           |                           | 10.9                        | 27.7      | 1.4      | 7.7  | 106.5 |
|                                   | <63 µm    | 69.4                       |                              |                           |                           | 12.3                        | 27.8      | 0.9      | 6.1  | 116.5 |
|                                   | CLEAN     |                            |                              |                           |                           |                             |           |          |      |       |

**CH4-9:** X-ray fluorescence (XRF) analyses on chemical composition of rocks (divided in major and trace elements), for both grain size fractions (63-125 µm and <63 µm). International standard OU-6 was used to test for accuracy, precision and limits of detection (LOD). Precision is given in percentage relative standard deviation (%RSD), which is expressed as the standard deviation of the mean as a percentage of the mean.

#### MAJOR ELEMENTS: DATA (%)

| ID                             | JdF mix 1 |        | JdF mix 2 |        | JdF mix 3 |        | CD80WP132 |        | G1        |        |              |
|--------------------------------|-----------|--------|-----------|--------|-----------|--------|-----------|--------|-----------|--------|--------------|
|                                | 63-125 µm | <63 µm | 63-125 µm | <63 µm | 63-125 µm | <63 µm | 63-125 µm | <63 µm | 63-125 µm | <63 µm | <63 µm CLEAN |
| SiO <sub>2</sub>               | 49.56     | 48.32  | 48.60     | 49.64  | 50.06     | 49.55  | 49.00     | 50.86  | 44.89     | 47.36  | 46.92        |
| TiO <sub>2</sub>               | 1.93      | 2.03   | 1.66      | 1.60   | 1.70      | 1.95   | 1.24      | 1.05   | 0.09      | 0.10   | 0.10         |
| Al <sub>2</sub> O <sub>3</sub> | 13.56     | 13.33  | 13.66     | 14.27  | 13.74     | 14.18  | 12.54     | 13.65  | 17.20     | 14.38  | 15.67        |
| Fe <sub>2</sub> O <sub>3</sub> | 11.72     | 12.41  | 11.19     | 10.63  | 11.63     | 12.23  | 14.45     | 12.55  | 5.54      | 6.44   | 6.08         |
| MnO                            | 0.23      | 0.24   | 0.20      | 0.19   | 0.20      | 0.21   | 0.22      | 0.19   | 0.11      | 0.13   | 0.12         |
| MgO                            | 6.47      | 6.82   | 7.40      | 7.18   | 6.79      | 6.35   | 7.65      | 7.57   | 11.08     | 13.17  | 11.77        |
| CaO                            | 12.39     | 12.88  | 13.88     | 13.41  | 12.86     | 12.84  | 12.88     | 12.29  | 17.09     | 14.07  | 15.38        |
| Na <sub>2</sub> O              | 2.35      | 2.20   | 2.15      | 2.09   | 2.12      | 2.05   | 1.70      | 1.51   | 0.95      | 1.08   | 0.89         |
| K <sub>2</sub> O               | 0.20      | 0.21   | 0.12      | 0.11   | 0.18      | 0.13   | 0.02      | 0.02   | 0.04      | 0.05   | 0.05         |
| P <sub>2</sub> O <sub>5</sub>  | 0.15      | 0.12   | 0.07      | 0.05   | 0.06      | 0.06   | 0.05      | 0.06   | <0.01     | <0.01  | <0.01        |
| SO <sub>3</sub>                | 0.10      | 0.10   | 0.13      | 0.14   | 0.08      | 0.11   | 0.24      | 0.21   | <0.03     | <0.03  | <0.03        |
| LOI                            | 1.33      | 1.33   | 0.93      | 0.69   | 0.58      | 0.35   | -0.18     | -0.38  | 2.70      | 2.90   | 2.70         |
| TOTAL                          | 99.99     | 99.99  | 99.99     | 100.00 | 100.00    | 100.01 | 99.82     | 99.58  | 99.66     | 99.64  | 99.66        |

#### TRACE ELEMENTS: DATA (ppm)

| ID | JdF mix 1 |        | JdF mix 2 |        | JdF mix 3 |        | CD80WP132 |        | G1        |        |              |
|----|-----------|--------|-----------|--------|-----------|--------|-----------|--------|-----------|--------|--------------|
|    | 63-125 µm | <63 µm | 63-125 µm | <63 µm | 63-125 µm | <63 µm | 63-125 µm | <63 µm | 63-125 µm | <63 µm | <63 µm CLEAN |
| V  | 380       | 372    | 371       | 354    | 375       | 379    | 347       | 383    | 140       | 158    | 169          |
| Cr | 250       | 247    | 306       | 264    | 276       | 201    | 210       | 246    | 488       | 584    | 525          |
| Co | 51        | 44     | 47        | 47     | 58        | 47     | 53        | 59     | 24        | 36     | 31           |
| Ni | 62        | 70     | 62        | 58     | 48        | 50     | 78        | 76     | 96        | 120    | 108          |
| Cu | 53        | 54     | 54        | 47     | 39        | 40     | 113       | 114    | 27        | 46     | 20           |
| Zn | 108       | 108    | 100       | 98     | 105       | 101    | 102       | 101    | 27        | 33     | 29           |
| Ga | 17        | 17     | 19        | 18     | 18        | 19     | 17        | 15     | 9         | 9      | 8            |
| Ge | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <1        | <1     | <1           |
| As | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <1        | <1     | <1           |
| Se | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <1        | <1     | <1           |
| Br | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <1        | <1     | <1           |
| Rb | 4         | 4      | 2         | 3      | 5         | 4      | 2         | 1      | 1         | 2      | 2            |
| Sr | 120       | 123    | 127       | 131    | 126       | 125    | 68        | 72     | 50        | 45     | 45           |
| Y  | 42        | 41     | 32        | 31     | 32        | 28     | 30        | 26     | 3         | 3      | 3            |
| Zr | 127       | 121    | 89        | 83     | 86        | 76     | 62        | 58     | 7         | 7      | 7            |
| Nb | 5         | 5      | <1        | 1      | 5         | 4      | 3         | <2     | 2         | 1      | 1            |
| Mo | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <2        | <2     | <2           |
| Ag | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <2        | <2     | <2           |
| Cd | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <2        | <2     | <2           |
| In | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <3           |
| Sn | <3        | <3     | <3        | <3     | <3        | <3     | <3        | 2      | <3        | <3     | <3           |
| Sb | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <3           |
| Ba | 68        | 73     | 64        | 71     | 50        | 100    | 98        | 56     | 5         | 9      | 7            |
| La | 8         | <5     | <5        | 6      | 5         | <5     | 5         | <5     | <3        | <3     | <3           |
| Ce | 19        | 10     | 12        | 13     | 12        | 9      | 11        | 10     | 5         | 4      | 5            |
| Pb | <3        | <3     | <3        | <3     | <3        | <3     | 6         | 2      | 1         | 1      | 1            |
| Th | <3        | <3     | <3        | <3     | 3         | <3     | <3        | 1      | <2        | <2     | <2           |
| U  | <3        | <3     | <3        | <3     | <3        | <3     | <3        | <3     | <2        | <2     | <2           |

**STANDARD OU-6 ANALYSES:**

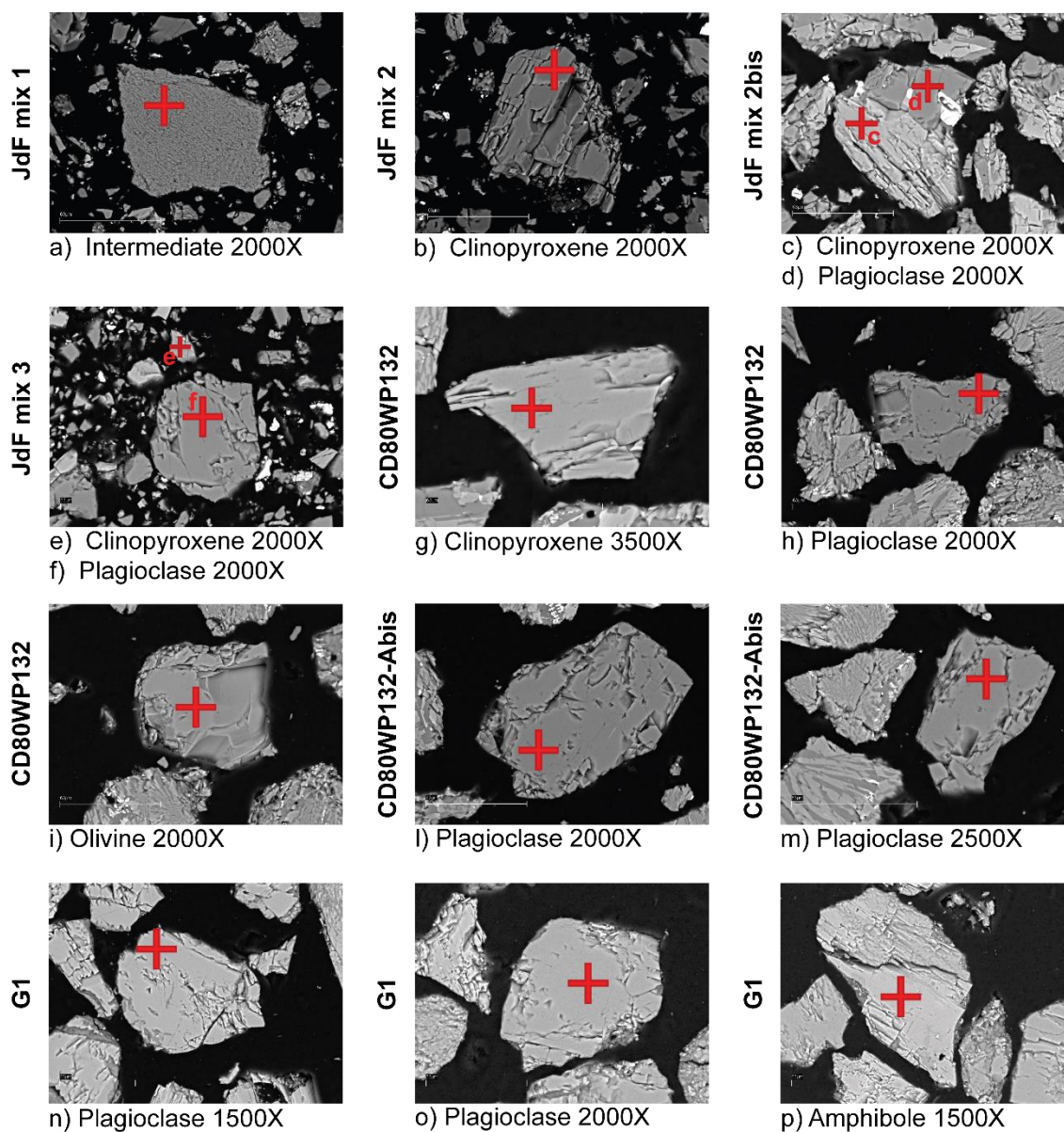
| MAJOR ELEMENTS                 | Certified value (%) | Average N=36 (%) | Accuracy (%) | % RSD | LOD  |
|--------------------------------|---------------------|------------------|--------------|-------|------|
| SiO <sub>2</sub>               | 57.35               | 57.33            | 0.41         | 0.20  | 0.35 |
| TiO <sub>2</sub>               | 0.99                | 0.99             | 1.65         | 0.82  | 0.02 |
| Al <sub>2</sub> O <sub>3</sub> | 20.45               | 20.34            | 0.58         | 0.29  | 0.10 |
| Fe <sub>2</sub> O <sub>3</sub> | 8.99                | 8.99             | 0.99         | 0.50  | 0.13 |
| MnO                            | 0.28                | 0.30             | 2.27         | 1.06  | 0.01 |
| MgO                            | 2.41                | 2.37             | 2.73         | 1.38  | 0.10 |
| CaO                            | 0.74                | 0.76             | 1.96         | 0.96  | 0.02 |
| Na <sub>2</sub> O              | 1.78                | 1.78             | 4.72         | 2.36  | 0.13 |
| K <sub>2</sub> O               | 3.04                | 3.01             | 1.43         | 0.72  | 0.07 |
| P <sub>2</sub> O <sub>5</sub>  | 0.12                | 0.12             | 10.82        | 5.39  | 0.02 |

| TRACE ELEMENTS | Certified value (ppm) | Average N=10 (ppm) | Accuracy (%) | % RSD | LOD  |
|----------------|-----------------------|--------------------|--------------|-------|------|
| V              | 129                   | 130                | 3.27         | 1.62  | 6.32 |
| Cr             | 71                    | 70.8               | 4.08         | 2.04  | 4.34 |
| Co             | 29                    | 29.4               | 12.55        | 6.19  | 5.46 |
| Ni             | 40                    | 40.0               | 6.38         | 3.19  | 3.83 |
| Cu             | 40                    | 39.5               | 5.24         | 2.65  | 3.14 |
| Zn             | 111                   | 113                | 6.16         | 3.03  | 10.2 |
| Ga             | 24                    | 23.9               | 4.92         | 2.48  | 1.77 |
| Ge             | #                     | #                  | #            | #     | #    |
| As             | 13                    | 13.0               | 11.18        | 5.60  | 2.18 |
| Se             | #                     | #                  | #            | #     | #    |
| Br             | #                     | #                  | #            | #     | #    |
| Rb             | 120                   | 121                | 1.34         | 0.66  | 2.41 |
| Sr             | 132                   | 129                | 3.10         | 1.58  | 6.13 |
| Y              | 28                    | 28.0               | 10.33        | 5.16  | 4.34 |
| Zr             | 174                   | 176                | 3.02         | 1.50  | 7.89 |
| Nb             | 14                    | 12.4               | 7.35         | 4.16  | 1.54 |
| Mo             | 1                     | 1.30               | 10.11        | 3.90  | 0.15 |
| Ag             | #                     | #                  | #            | #     | #    |
| Cd             | #                     | #                  | #            | #     | #    |
| In             | #                     | #                  | #            | #     | #    |
| Sn             | 3                     | 2.71               | 20.00        | 11.08 | 0.90 |
| Sb             | 1                     | 1.16               | 35.68        | 7.69  | 0.27 |
| Ba             | 480                   | 475                | 4.89         | 2.47  | 35.2 |
| La             | 33                    | 32.3               | 21.87        | 11.18 | 10.8 |
| Ce             | 77                    | 77.7               | 8.33         | 4.12  | 9.62 |
| Pb             | 28                    | 29.9               | 14.89        | 6.97  | 6.26 |
| Th             | 12                    | 13.0               | 3.41         | 1.58  | 0.61 |
| U              | 2                     | 2.18               | 50.19        | 23.03 | 1.51 |

**CH4-10:** Summary of SEM-EDS analyses on grains in ground samples (63-125  $\mu\text{m}$ ), before each experiments.

The red crosses on SEM-EDS images indicates the analysed spot. Images, data, and correspondent analytical error (standard deviation,  $\sigma$ ) are shown for each sample used in the dissolution experiments.

Accuracy and precision for SEM-EDS analyses have been calibrated with the standard glass sample BIR1G (See Appendix CH4-6).



# DATA FOR OXIDES ANALYSES (%)

|   | Sample             | Mineral | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | MnO  | MgO   | CaO   | Na <sub>2</sub> O | Cr <sub>2</sub> O <sub>3</sub> | TOT    |
|---|--------------------|---------|------------------|------------------|--------------------------------|--------------------------------|------|-------|-------|-------------------|--------------------------------|--------|
| a | JdF mix 1          | Int     | 48.95            | 2.90             | 12.76                          | 12.65                          |      | 8.81  | 12.58 | 2.20              |                                | 100.84 |
| b | JdF mix 2          | Cpx     | 50.99            | 0.75             | 2.70                           | 8.32                           |      | 17.09 | 18.51 | 0.29              |                                | 98.65  |
| c | JdF mix 2bis       | Cpx     | 50.62            | 1.67             | 4.37                           | 12.62                          | 0.35 | 16.79 | 16.44 |                   |                                | 102.85 |
| d | JdF mix 2bis       | Plg     | 54.32            |                  | 28.08                          | 1.23                           |      |       | 10.74 | 5.13              |                                | 99.50  |
| e | JdF mix 3          | Cpx     | 51.85            | 0.96             | 3.69                           | 8.48                           |      | 18.35 | 17.86 |                   |                                | 101.17 |
| f | JdF mix 3          | Plg     | 53.31            |                  | 28.91                          | 1.01                           |      |       | 11.89 | 4.89              |                                | 100.01 |
| g | CD80WP132          | Cpx     | 51.27            | 0.31             | 2.45                           | 6.95                           |      | 17.65 | 19.39 | 0.30              | 0.35                           | 98.66  |
| h | CD80WP132          | Plg     | 45.46            |                  | 32.49                          | 0.80                           |      | 0.25  | 16.17 | 2.11              |                                | 97.27  |
| i | CD80WP132          | Ol      | 37.29            |                  |                                | 21.93                          | 0.29 | 40.04 | 0.38  |                   |                                | 99.93  |
| l | CD80WP132-<br>Abis | Plg     | 48.82            |                  | 31.66                          | 0.85                           |      | 0.31  | 14.91 | 2.91              |                                | 99.47  |
| m | CD80WP132-<br>Abis | Plg     | 48.62            |                  | 32.45                          | 0.71                           |      | 0.29  | 15.65 | 2.83              |                                | 100.55 |
| n | G1                 | Plg-An  | 42.82            |                  | 35.57                          | 0.55                           |      |       | 19.02 | 0.56              |                                | 98.51  |
| o | G1                 | Plg-An  | 42.65            |                  | 35.44                          | 0.65                           |      |       | 19.29 | 0.64              |                                | 98.68  |
| p | G1                 | Amph    | 52.77            | 0.30             | 1.13                           | 6.21                           |      | 17.14 | 21.63 |                   |                                | 99.18  |

# ERROR (σ) FOR OXIDES ANALYSES (%)

|   | Sample             | Mineral | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | MnO  | MgO  | CaO  | Na <sub>2</sub> O | Cr <sub>2</sub> O <sub>3</sub> |
|---|--------------------|---------|------------------|------------------|--------------------------------|--------------------------------|------|------|------|-------------------|--------------------------------|
| a | JdF mix 1          | Int     | 0.29             | 0.11             | 0.16                           | 0.20                           |      | 0.14 | 0.13 | 0.10              |                                |
| b | JdF mix 2          | Cpx     | 0.29             | 0.08             | 0.10                           | 0.17                           |      | 0.17 | 0.15 | 0.07              |                                |
| c | JdF mix 2bis       | Cpx     | 0.31             | 0.10             | 0.12                           | 0.20                           | 0.08 | 0.18 | 0.14 |                   |                                |
| d | JdF mix 2bis       | Plg     | 0.31             |                  | 0.21                           | 0.10                           |      |      | 0.12 | 0.12              |                                |
| e | JdF mix 3          | Cpx     | 0.31             | 0.09             | 0.11                           | 0.18                           |      | 0.18 | 0.15 |                   |                                |
| f | JdF mix 3          | Plg     | 0.31             |                  | 0.21                           | 0.10                           |      |      | 0.13 | 0.11              |                                |
| g | CD80WP132          | Cpx     | 0.29             | 0.08             | 0.10                           | 0.16                           |      | 0.17 | 0.15 | 0.07              | 0.08                           |
| h | CD80WP132          | Plg     | 0.29             |                  | 0.22                           | 0.10                           |      | 0.06 | 0.14 | 0.09              |                                |
| i | CD80WP132          | Ol      | 0.28             |                  |                                | 0.25                           | 0.07 | 0.25 | 0.05 |                   |                                |
| l | CD80WP132-<br>Abis | Plg     | 0.30             |                  | 0.22                           | 0.10                           |      | 0.07 | 0.14 | 0.10              |                                |
| m | CD80WP132-<br>Abis | Plg     | 0.30             |                  | 0.22                           | 0.09                           |      | 0.07 | 0.14 | 0.10              |                                |
| n | G1                 | Plg-An  | 0.29             |                  | 0.23                           | 0.09                           |      |      | 0.15 | 0.07              |                                |
| o | G1                 | Plg-An  | 0.29             |                  | 0.23                           | 0.09                           |      |      | 0.15 | 0.07              |                                |
| p | G1                 | Amph    | 0.31             | 0.08             | 0.09                           | 0.16                           |      | 0.17 | 0.16 |                   |                                |

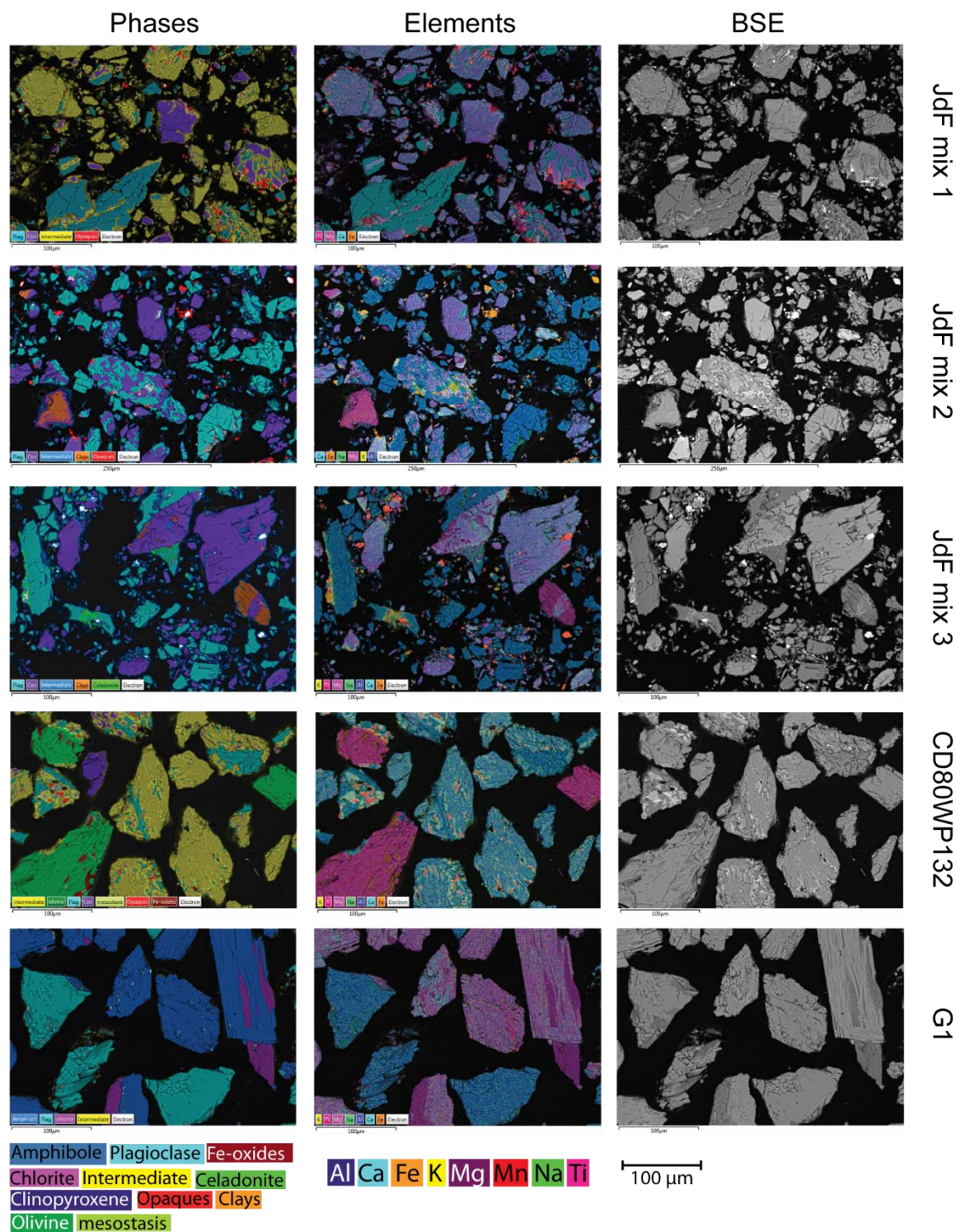


**CH4-11:** Summary of SEM analyses on ground samples (63-125  $\mu\text{m}$ ), before and after each experiments.

SEM images are shown divided for each rock in BSE-back scatter electron, elemental line, and mineral phase mode. SEM-EDS analyses and correspondent analytical error (standard deviation,  $\sigma$ ) are shown for both bulk rock and single phases. To note: the total for SEM-EDS analyses is forced to be 100 % from the software.

Accuracy and precision for SEM-EDS analyses have been calibrated with the standard glass sample BIR1G (See Appendix CH4-6).

**INITIAL SAMPLES (before experiments):**



## DATA FOR OXIDES ANALYSES (%)

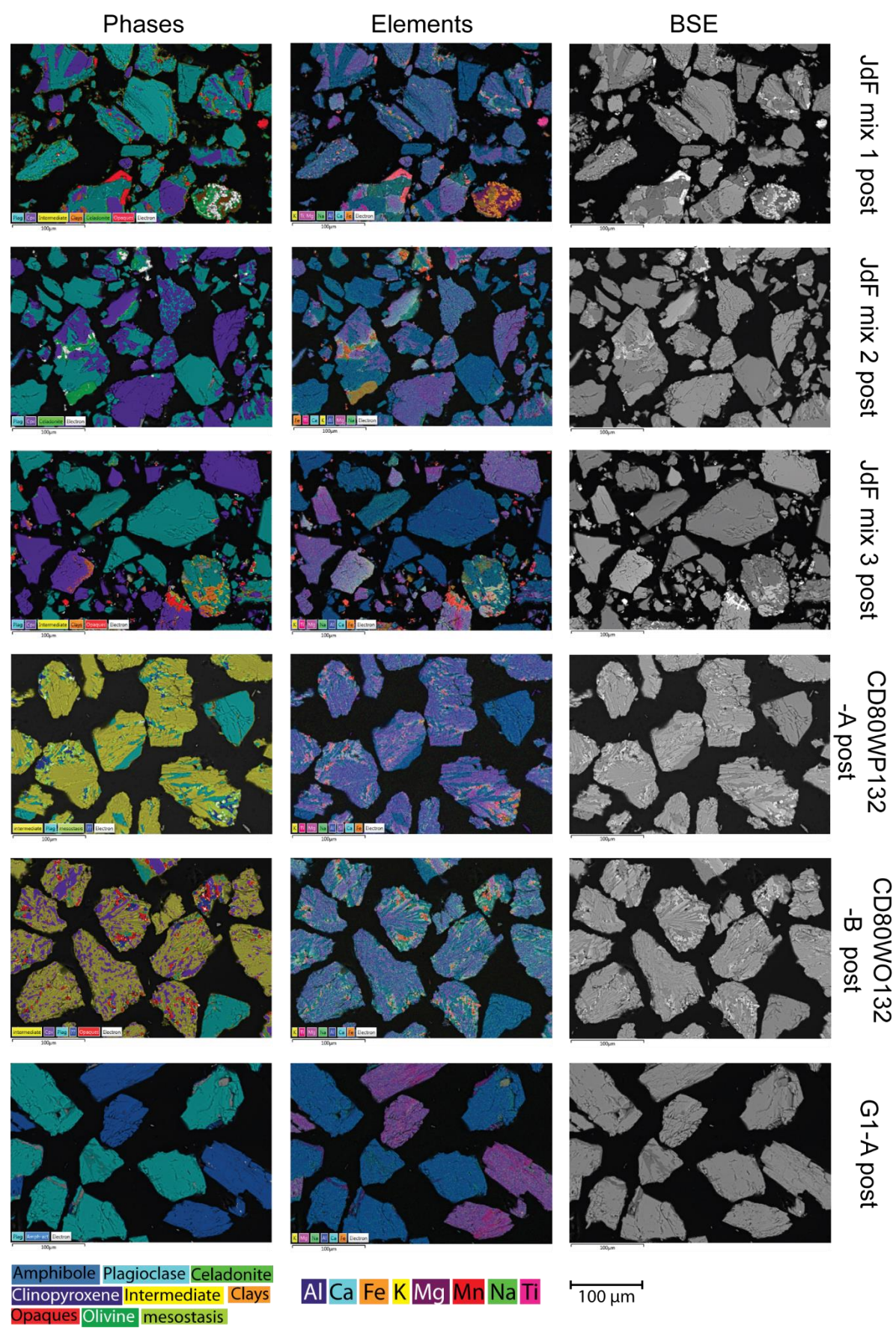
|                         | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | FeO   | MnO  | MgO   | CaO   | Na <sub>2</sub> O | K <sub>2</sub> O | P <sub>2</sub> O <sub>5</sub> | SO <sub>3</sub> |
|-------------------------|------------------|------------------|--------------------------------|--------------------------------|-------|------|-------|-------|-------------------|------------------|-------------------------------|-----------------|
| <b>JdF mix 1</b>        |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |
| Bulk rock               | 50.80            | 1.85             | 16.91                          |                                | 8.59  | 0.15 | 6.70  | 11.26 | 3.32              | 0.25             |                               | 0.17            |
| Plagioclase-labradorite | 53.02            |                  | 28.30                          |                                | 1.53  |      | 0.73  | 11.20 | 5.09              | 0.13             |                               |                 |
| Clinopyroxene           | 50.66            | 1.43             | 4.21                           |                                | 10.61 | 0.25 | 16.88 | 15.46 | 0.51              |                  |                               |                 |
| Intermediate            | 51.15            | 1.82             | 14.83                          |                                | 9.67  | 0.17 | 7.51  | 11.15 | 3.09              | 0.31             | 0.12                          | 0.18            |
| Opaques                 | 15.46            | 21.00            | 5.02                           | 50.36                          |       | 1.32 | 2.47  | 2.35  | 1.63              | 0.38             |                               |                 |
| <b>JdF mix 2</b>        |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |
| Bulk rock               | 50.75            | 1.41             | 16.73                          |                                | 8.45  | 0.15 | 7.45  | 11.66 | 3.00              | 0.26             |                               | 0.14            |
| Plagioclase-labradorite | 53.13            |                  | 26.37                          | 2.77                           |       |      | 1.56  | 11.30 | 4.87              |                  |                               |                 |
| Clinopyroxene           | 51.59            | 1.15             | 4.80                           |                                | 11.47 |      | 15.76 | 14.50 | 0.73              |                  |                               |                 |
| Intermediate            | 52.98            | 1.58             | 15.00                          |                                | 9.25  |      | 7.12  | 11.12 | 2.57              | 0.37             |                               |                 |
| Clays-saponite          | 55.6             |                  | 3.14                           |                                | 17.07 |      | 20.27 | 2.02  | 0.78              | 1.11             |                               |                 |
| Opaques                 | 17.44            | 17.83            | 4.83                           | 52.23                          |       |      | 3.8   | 2.68  | 0.85              | 0.34             |                               |                 |
| <b>JdF mix 3</b>        |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |
| Bulk rock               | 51.16            | 1.46             | 14.86                          |                                | 10.12 | 0.14 | 7.63  | 10.95 | 3.29              | 0.29             |                               | 0.10            |
| Plagioclase-labradorite | 55.07            | 0.17             | 27.07                          |                                | 1.38  |      | 0.49  | 9.50  | 6.17              | 0.15             |                               |                 |
| Clinopyroxene           | 50.76            | 1.34             | 3.34                           |                                | 13.26 | 0.26 | 15.36 | 14.97 | 0.60              | 0.10             |                               |                 |
| Intermediate            | 50.13            | 2.47             | 16.30                          |                                | 12.39 |      | 5.13  | 9.55  | 3.34              | 0.51             |                               | 0.17            |
| Clays - saponite        | 54.98            | 0.34             | 4.71                           |                                | 16.26 |      | 20.15 | 1.83  | 1.09              | 0.62             |                               |                 |
| Celadonite              | 59.20            | 1.05             | 8.97                           |                                | 16.20 |      | 4.23  | 3.42  | 4.06              | 1.88             | 0.99                          |                 |
| <b>CD80WP132</b>        |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |
| Bulk rock               | 47.09            | 0.93             | 11.77                          |                                | 12.93 | 0.17 | 16.40 | 8.32  | 2.06              | 0.06             |                               | 0.27            |
| Plagioclase-labradorite | 52.66            | 0.34             | 26.97                          |                                | 2.67  |      | 0.95  | 11.45 | 4.70              | 0.07             |                               | 0.20            |
| Intermediate            | 51.23            | 0.98             | 14.23                          |                                | 10.71 | 0.17 | 7.81  | 11.96 | 2.56              | 0.07             |                               | 0.28            |
| Olivine                 | 39.10            |                  |                                |                                | 17.57 | 0.22 | 42.53 | 0.39  | 0.18              |                  |                               |                 |
| Clinopyroxene           | 51.51            | 0.62             | 3.77                           |                                | 9.30  | 0.24 | 17.52 | 16.56 | 0.47              |                  |                               |                 |
| Mesostasis              | 45.56            | 4.90             | 8.63                           |                                | 27.59 | 0.25 | 3.00  | 6.38  | 2.37              | 0.20             | 0.24                          | 0.88            |
| Opaques                 | 26.07            | 11.94            | 6.56                           | 50.35                          |       |      |       | 1.76  | 2.28              |                  |                               | 1.05            |
| Iddingsite              | 15.15            |                  |                                | 69.26                          |       |      | 14.00 | 1.59  |                   |                  |                               |                 |
| <b>G1</b>               |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |
| Bulk rock               | 49.69            | 0.09             | 17.15                          |                                | 5.73  | 0.10 | 13.80 | 12.64 | 0.71              | 0.08             |                               |                 |
| Plagioclase-anorthite   | 44.31            |                  | 35.92                          |                                | 0.58  |      | 0.30  | 17.91 | 0.97              |                  |                               |                 |
| Amphibole-actinolite    | 54.83            |                  | 4.23                           |                                | 8.14  | 0.17 | 20.11 | 11.78 | 0.54              | 0.08             |                               |                 |
| Chlorite                | 36.32            |                  | 20.74                          | 11.51                          |       |      | 30.13 | 1.10  | 0.20              |                  |                               |                 |
| Intermediate            | 54.15            |                  | 17.34                          |                                | 5.06  |      | 11.07 | 11.26 | 1.12              |                  |                               |                 |

## ERROR (σ) FOR OXIDES ANALYSES (%)

|                         | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | FeO  | MnO  | MgO  | CaO  | Na <sub>2</sub> O | K <sub>2</sub> O | P <sub>2</sub> O <sub>5</sub> | SO <sub>3</sub> |
|-------------------------|------------------|------------------|--------------------------------|--------------------------------|------|------|------|------|-------------------|------------------|-------------------------------|-----------------|
| <b>JdF mix 1</b>        |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |
| Bulk rock               | 0.02             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.02            |
| Plagioclase-labradorite | 0.10             |                  | 0.07                           |                                | 0.04 |      | 0.03 | 0.04 | 0.04              | 0.01             |                               |                 |
| Clinopyroxene           | 0.15             | 0.05             | 0.07                           |                                | 0.09 | 0.04 | 0.09 | 0.07 | 0.05              |                  |                               |                 |
| Intermediate            | 0.07             | 0.02             | 0.04                           |                                | 0.04 | 0.02 | 0.03 | 0.03 | 0.03              | 0.01             | 0.02                          | 0.02            |
| Opaques                 | 0.29             | 0.30             | 0.21                           | 0.43                           |      | 0.15 | 0.20 | 0.11 | 0.21              | 0.07             |                               |                 |
| <b>JdF mix 2</b>        |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |
| Bulk rock               | 0.06             | 0.02             | 0.03                           |                                | 0.03 | 0.02 | 0.03 | 0.02 | 0.02              | 0.01             |                               | 0.02            |
| Plagioclase-labradorite | 0.08             |                  | 0.06                           | 0.04                           |      |      | 0.02 | 0.03 | 0.03              |                  |                               |                 |
| Clinopyroxene           | 0.10             | 0.03             | 0.04                           |                                | 0.06 |      | 0.06 | 0.05 | 0.03              |                  |                               |                 |
| Intermediate            | 0.14             | 0.06             | 0.09                           |                                | 0.08 |      | 0.07 | 0.06 | 0.06              | 0.03             |                               |                 |
| Clays-saponite          | 0.36             |                  | 0.14                           |                                | 0.24 |      | 0.22 | 0.09 | 0.11              | 0.07             |                               |                 |
| Opaques                 | 0.29             | 0.27             | 0.20                           | 0.41                           |      |      | 0.20 | 0.11 | 0.20              | 0.07             |                               |                 |
| <b>JdF mix 3</b>        |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |
| Bulk rock               | 0.06             | 0.02             | 0.03                           |                                | 0.03 | 0.02 | 0.03 | 0.02 | 0.02              | 0.01             |                               | 0.02            |
| Plagioclase-labradorite | 0.10             | 0.03             | 0.07                           |                                | 0.03 |      | 0.03 | 0.04 | 0.04              | 0.01             |                               |                 |
| Clinopyroxene           | 0.09             | 0.03             | 0.04                           |                                | 0.06 | 0.02 | 0.05 | 0.04 | 0.03              | 0.01             |                               |                 |
| Intermediate            | 0.11             | 0.04             | 0.07                           |                                | 0.07 |      | 0.05 | 0.04 | 0.05              | 0.02             |                               | 0.04            |
| Clays - saponite        | 0.33             | 0.09             | 0.15                           |                                | 0.21 |      | 0.21 | 0.08 | 0.12              | 0.05             |                               |                 |
| Celadonite              | 0.37             | 0.11             | 0.18                           |                                | 0.23 |      | 0.15 | 0.10 | 0.16              | 0.07             | 0.15                          |                 |
| <b>CD80WP132</b>        |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |
| Bulk rock               | 0.04             | 0.01             | 0.03                           |                                | 0.03 | 0.01 | 0.03 | 0.02 | 0.02              | 0.01             |                               | 0.02            |
| Plagioclase-labradorite | 0.04             | 0.04             | 0.10                           |                                | 0.05 |      | 0.04 | 0.06 | 0.06              | 0.02             |                               | 0.04            |
| Intermediate            | 0.06             | 0.02             | 0.04                           |                                | 0.04 | 0.02 | 0.03 | 0.03 | 0.02              | 0.01             |                               | 0.02            |
| Olivine                 | 0.08             |                  |                                |                                | 0.06 | 0.02 | 0.07 | 0.02 | 0.03              |                  |                               |                 |
| Clinopyroxene           | 0.22             | 0.07             | 0.10                           |                                | 0.13 | 0.06 | 0.14 | 0.11 | 0.07              |                  |                               |                 |
| Mesostasis              | 0.20             | 0.09             | 0.11                           |                                | 0.16 | 0.06 | 0.08 | 0.07 | 0.09              | 0.03             | 0.07                          | 0.07            |
| Opaques                 | 0.41             | 0.29             | 0.26                           | 0.51                           |      |      |      | 0.12 | 0.25              |                  |                               | 0.18            |
| Iddingsite              | 0.61             |                  |                                | 1.04                           |      |      | 0.63 | 0.26 |                   |                  |                               |                 |
| <b>G1</b>               |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |
| Bulk rock               | 0.05             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.03 | 0.02 | 0.02              | 0.01             |                               |                 |
| Plagioclase-anorthite   | 0.08             |                  | 0.07                           |                                | 0.03 |      | 0.02 | 0.04 | 0.03              |                  |                               |                 |
| Amphibole-actinolite    | 0.07             |                  | 0.03                           |                                | 0.04 | 0.01 | 0.04 | 0.03 | 0.02              | 0.01             |                               |                 |
| Chlorite                | 0.18             |                  | 0.15                           | 0.13                           |      |      | 0.14 | 0.04 | 0.06              |                  |                               |                 |
| Intermediate            | 0.57             |                  | 0.38                           |                                | 0.31 |      | 0.32 | 0.26 | 0.28              |                  |                               |                 |



**SAMPLES AFTER FIRST SET OF EXPERIMENTS:**



## DATA FOR OXIDES ANALYSES (%)

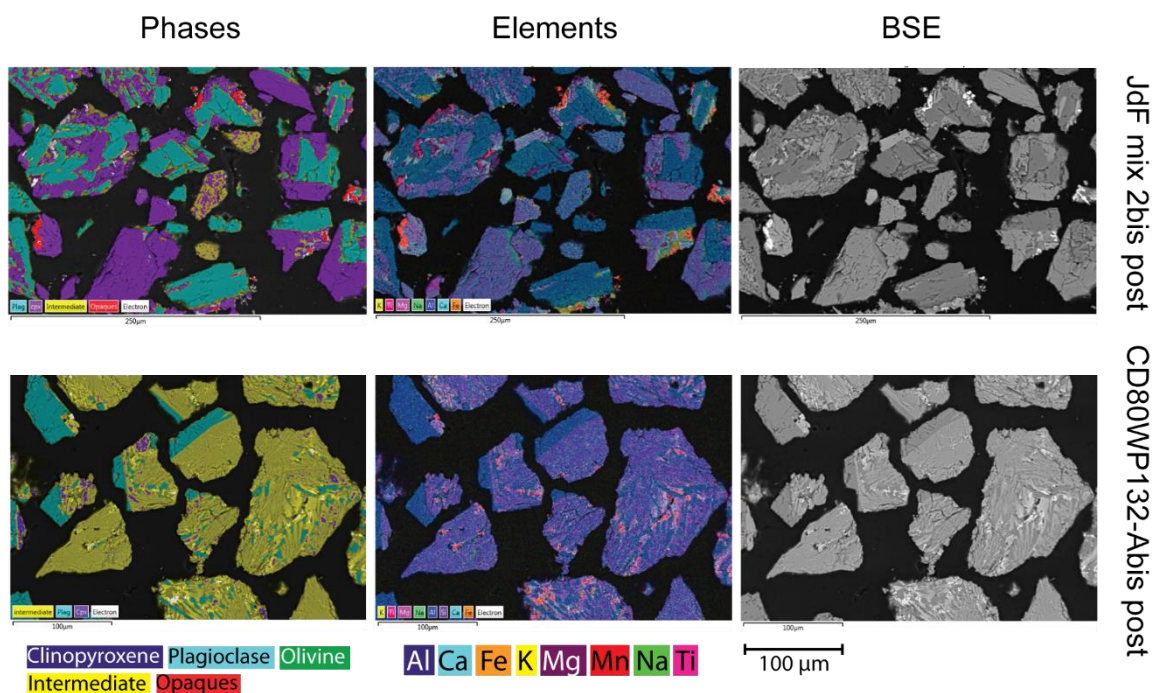
|                         | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | FeO   | MnO  | MgO   | CaO   | Na <sub>2</sub> O | K <sub>2</sub> O | P <sub>2</sub> O <sub>5</sub> | SO <sub>3</sub> | Cr <sub>2</sub> O <sub>3</sub> |
|-------------------------|------------------|------------------|--------------------------------|--------------------------------|-------|------|-------|-------|-------------------|------------------|-------------------------------|-----------------|--------------------------------|
| <b>JdF mix 1 post</b>   |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |                                |
| Bulk rock               | 50.52            | 2.14             | 15.33                          |                                | 1.01  | 0.18 | 6.89  | 10.09 | 3.45              | 0.26             |                               | 0.13            |                                |
| Plagioclase-labradorite | 53.43            | 1.04             | 19.84                          |                                | 6.00  |      | 4.53  | 10.28 | 4.56              | 0.20             | 0.13                          |                 |                                |
| Clinopyroxene           | 50.20            | 1.74             | 4.62                           |                                | 12.66 | 0.31 | 15.02 | 14.75 | 0.72              |                  |                               |                 |                                |
| Intermediate            | 51.73            | 2.41             | 15.06                          |                                | 11.19 |      | 6.07  | 10.27 | 2.84              | 0.42             |                               |                 |                                |
| Clays-saponite          | 53.92            | 0.40             | 3.29                           |                                | 16.64 |      | 22.53 | 2.90  |                   | 0.32             |                               |                 |                                |
| Celadonite              | 47.00            | 3.13             | 7.26                           |                                | 23.75 | 0.32 | 9.11  | 6.68  | 1.61              | 1.12             |                               |                 |                                |
| Iddingsite              | 26.54            | 0.31             | 1.89                           | 64.15                          |       |      | 5.45  | 0.72  |                   | 0.93             |                               |                 |                                |
| Opagues                 | 13.67            | 22.73            | 5.07                           | 50.77                          |       | 1.37 | 1.95  | 2.09  | 1.49              | 0.21             | 0.64                          |                 |                                |
| <b>JdF mix 2 post</b>   |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |                                |
| Bulk rock               | 50.80            | 1.12             | 17.60                          |                                | 7.13  | 0.14 | 7.48  | 12.37 | 3.05              | 0.24             |                               | 0.06            |                                |
| Plagioclase-labradorite | 52.85            | 0.32             | 27.07                          |                                | 2.19  |      | 1.36  | 11.23 | 4.85              | 0.13             |                               |                 |                                |
| Clinopyroxene           | 50.85            | 1.25             | 6.26                           |                                | 9.72  | 0.21 | 15.42 | 15.07 | 0.94              | 0.10             |                               |                 | 0.17                           |
| Celadonite              | 52.34            | 1.54             | 8.29                           |                                | 21.73 | 0.32 | 4.89  | 5.47  | 1.95              | 2.98             | 0.49                          |                 |                                |
| <b>JdF mix 3 post</b>   |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |                                |
| Bulk rock               | 51.32            | 1.17             | 18.1                           |                                | 7.74  | 0.15 | 5.98  | 11.69 | 3.63              | 0.23             |                               |                 |                                |
| Plagioclase-labradorite | 53.95            | 0.16             | 27.91                          |                                | 1.33  |      | 0.51  | 10.49 | 5.50              | 0.15             |                               |                 |                                |
| Clinopyroxene           | 51.20            | 1.14             | 2.90                           |                                | 12.40 | 0.29 | 15.82 | 15.69 | 0.48              | 0.08             |                               |                 |                                |
| Intermediate            | 52.58            | 1.03             | 12.17                          |                                | 14.73 | 0.32 | 5.72  | 7.27  | 3.79              | 1.27             | 1.14                          |                 |                                |
| Clays                   | 47.08            | 1.36             | 1.98                           |                                | 28.75 | 0.68 | 6.93  | 11.84 | 0.72              | 0.18             | 0.48                          |                 |                                |
| Opagues                 | 6.20             | 23.21            | 2.77                           | 63.86                          |       | 1.21 | 1.66  | 0.83  |                   | 0.28             |                               |                 |                                |
| <b>CD80WP132-A post</b> |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |                                |
| Bulk rock               | 50.89            | 1.22             | 15.74                          |                                | 10.99 | 0.15 | 6.56  | 11.41 | 2.67              | 0.08             |                               | 0.29            |                                |
| Plagioclase-labradorite | 51.52            | 0.20             | 28.97                          |                                | 2.01  |      | 0.67  | 12.57 | 3.99              | 0.07             |                               |                 |                                |
| Intermediate            | 51.14            | 0.99             | 13.44                          |                                | 10.96 | 0.19 | 8.64  | 12.12 | 2.32              |                  |                               | 0.21            |                                |
| Mesostasis              | 45.86            | 4.64             | 8.79                           |                                | 27.59 | 0.29 | 2.68  | 6.43  | 2.61              | 0.28             |                               | 0.83            |                                |
| <b>CD80WP132-B post</b> |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |                                |
| Bulk rock               | 50.64            | 1.27             | 15.74                          |                                | 10.90 | 0.16 | 6.67  | 11.54 | 2.72              | 0.07             |                               | 0.28            |                                |
| Plagioclase-labradorite | 49.48            |                  | 32.16                          |                                | 0.78  |      | 0.31  | 13.99 | 3.29              |                  |                               |                 |                                |
| Clinopyroxene           | 50.24            | 1.02             | 7.88                           |                                | 13.50 | 0.28 | 12.91 | 12.99 | 1.17              |                  |                               |                 |                                |
| Intermediate            | 51.69            | 0.98             | 16.57                          |                                | 9.39  | 0.15 | 6.14  | 11.70 | 3.06              | 0.07             |                               | 0.26            |                                |
| Opagues                 | 36.76            | 7.93             | 7.95                           | 38.20                          |       | 0.30 | 1.34  | 3.66  | 2.59              | 0.17             |                               | 1.09            |                                |
| <b>G1-A post</b>        |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |                                |
| Bulk rock               | 48.26            | 3.23             | 25.12                          |                                | 3.23  |      | 8.05  | 14.49 | 0.77              | 0.08             |                               |                 |                                |
| Plagioclase-anorthite   | 44.11            |                  | 36.15                          |                                | 0.54  |      | 0.28  | 18.06 | 0.86              |                  |                               |                 |                                |
| Amphibole-actinolite    | 56.00            | 0.14             | 4.11                           |                                | 7.59  | 0.14 | 21.17 | 10.20 | 0.45              | 0.06             |                               |                 | 0.13                           |

## ERROR (σ) FOR OXIDES ANALYSES (%)

|                         | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | FeO  | MnO  | MgO  | CaO  | Na <sub>2</sub> O | K <sub>2</sub> O | P <sub>2</sub> O <sub>5</sub> | SO <sub>3</sub> | Cr <sub>2</sub> O <sub>3</sub> |
|-------------------------|------------------|------------------|--------------------------------|--------------------------------|------|------|------|------|-------------------|------------------|-------------------------------|-----------------|--------------------------------|
| <b>JdF mix 1 post</b>   |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |                                |
| Bulk rock               | 0.05             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.02            |                                |
| Plagioclase-labradorite | 0.06             | 0.02             | 0.04                           |                                | 0.03 |      | 0.02 | 0.03 | 0.03              | 0.01             | 0.02                          |                 |                                |
| Clinopyroxene           | 0.12             | 0.04             | 0.05                           |                                | 0.08 | 0.03 | 0.07 | 0.06 | 0.04              |                  |                               |                 |                                |
| Intermediate            | 0.16             | 0.07             | 0.10                           |                                | 0.10 |      | 0.07 | 0.07 | 0.03              |                  |                               |                 |                                |
| Clays-saponite          | 0.42             | 0.11             | 0.17                           |                                | 0.28 |      |      | 0.11 |                   | 0.07             |                               |                 |                                |
| Celadonite              | 0.24             | 0.09             | 0.12                           |                                | 0.18 | 0.07 | 0.12 | 0.08 | 0.09              | 0.05             |                               |                 |                                |
| Iddingsite              | 0.25             | 0.09             | 0.12                           | 0.31                           |      |      | 0.16 | 0.06 |                   | 0.06             |                               |                 |                                |
| Opagues                 | 0.19             | 0.21             | 0.14                           | 0.30                           |      | 0.10 | 0.12 | 0.07 | 0.14              | 0.05             | 0.11                          |                 |                                |
| <b>JdF mix 2 post</b>   |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |                                |
| Bulk rock               | 0.04             | 0.01             | 0.03                           |                                | 0.02 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.01            |                                |
| Plagioclase-labradorite | 0.06             | 0.02             | 0.04                           |                                | 0.02 |      | 0.02 | 0.03 | 0.03              | 0.01             |                               |                 |                                |
| Clinopyroxene           | 0.06             | 0.02             | 0.03                           |                                | 0.04 | 0.02 | 0.04 | 0.03 | 0.02              | 0.01             |                               |                 | 0.02                           |
| Celadonite              | 0.26             | 0.08             | 0.13                           |                                | 0.18 | 0.07 | 0.11 | 0.08 | 0.10              | 0.06             | 0.10                          |                 |                                |
| <b>JdF mix 3 post</b>   |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |                                |
| Bulk rock               | 0.05             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               |                 |                                |
| Plagioclase-labradorite | 0.06             | 0.02             | 0.05                           |                                | 0.02 |      | 0.02 | 0.03 | 0.03              | 0.01             |                               |                 |                                |
| Clinopyroxene           | 0.08             | 0.03             | 0.03                           |                                | 0.05 | 0.02 | 0.05 | 0.04 | 0.03              | 0.01             |                               |                 |                                |
| Intermediate            | 0.27             | 0.08             | 0.15                           |                                | 0.17 | 0.07 | 0.12 | 0.10 | 0.12              | 0.05             | 0.11                          |                 |                                |
| Clays                   | 0.32             | 0.11             | 0.12                           |                                | 0.26 | 0.10 | 0.16 | 0.14 | 0.12              | 0.05             | 0.12                          |                 |                                |
| Opagues                 | 0.24             | 0.35             | 0.21                           | 0.51                           |      | 0.17 | 0.21 | 0.10 |                   | 0.08             |                               |                 |                                |
| <b>CD80WP132-A post</b> |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |                                |
| Bulk rock               | 0.04             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.02            |                                |
| Plagioclase-labradorite | 0.11             | 0.03             | 0.08                           |                                | 0.04 |      | 0.03 | 0.05 | 0.04              | 0.02             |                               |                 |                                |
| Intermediate            | 0.05             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.03 | 0.02 | 0.02              |                  |                               | 0.02            |                                |
| Mesostasis              | 0.17             | 0.08             | 0.09                           |                                | 0.14 | 0.05 | 0.07 | 0.06 | 0.08              | 0.03             |                               | 0.06            |                                |
| <b>CD80WP132-B post</b> |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |                                |
| Bulk rock               | 0.04             | 0.01             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.02            |                                |
| Plagioclase-labradorite | 0.16             |                  | 0.12                           |                                | 0.05 |      | 0.04 | 0.07 | 0.06              |                  |                               |                 |                                |
| Clinopyroxene           | 0.09             | 0.03             | 0.05                           |                                | 0.06 | 0.03 | 0.05 | 0.04 | 0.03              |                  |                               |                 |                                |
| Intermediate            | 0.05             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.02            |                                |
| Opagues                 | 0.20             | 0.11             | 0.12                           | 0.21                           |      | 0.06 | 0.08 | 0.06 | 0.10              | 0.03             |                               | 0.08            |                                |
| <b>G1-A post</b>        |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |                                |
| Bulk rock               | 0.05             |                  | 0.04                           |                                | 0.02 |      | 0.02 | 0.02 | 0.02              | 0.01             |                               |                 |                                |
| Plagioclase-anorthite   | 0.06             |                  | 0.05                           |                                | 0.02 |      | 0.02 | 0.03 | 0.02              |                  |                               |                 |                                |
| Amphibole-actinolite    | 0.09             | 0.02             | 0.04                           |                                | 0.04 | 0.02 | 0.05 | 0.03 | 0.03              | 0.01             |                               |                 | 0.02                           |



## SAMPLES AFTER SECOND SET OF EXPERIMENTS:



## DATA FOR OXIDES ANALYSES (%)

|                            | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | FeO   | MnO  | MgO   | CaO   | Na <sub>2</sub> O | K <sub>2</sub> O | P <sub>2</sub> O <sub>5</sub> | SO <sub>3</sub> | Cr <sub>2</sub> O <sub>3</sub> |
|----------------------------|------------------|------------------|--------------------------------|--------------------------------|-------|------|-------|-------|-------------------|------------------|-------------------------------|-----------------|--------------------------------|
| <b>JdF mix 2bis post</b>   |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |                                |
| Bulk rock                  | 50.52            | 1.59             | 15.73                          |                                | 8.13  | 0.12 | 8.10  | 12.46 | 2.96              | 0.19             |                               | 0.10            | 0.11                           |
| Plagioclase-labradorite    | 53.57            | 0.18             | 27.79                          |                                | 1.34  |      | 0.68  | 10.96 | 5.37              | 0.10             |                               |                 |                                |
| Clinopyroxene              | 50.49            | 1.39             | 5.21                           |                                | 10.76 | 0.19 | 15.44 | 15.39 | 0.71              | 0.15             |                               | 0.09            | 0.18                           |
| Intermediate               | 52.75            | 1.04             | 19.34                          |                                | 6.90  |      | 4.89  | 10.82 | 3.99              | 0.27             |                               |                 |                                |
| Opaques                    | 7.02             | 22.13            | 2.80                           | 64.02                          |       | 1.04 | 1.89  | 0.83  |                   | 0.28             |                               |                 |                                |
| <b>CD80WP132-Abis post</b> |                  |                  |                                |                                |       |      |       |       |                   |                  |                               |                 |                                |
| Bulk rock                  | 50.69            | 1.23             | 16.25                          |                                | 10.65 | 0.14 | 6.44  | 11.50 | 2.72              | 0.08             |                               | 0.28            |                                |
| Plagioclase-labradorite    | 51.12            | 0.14             | 29.46                          |                                | 1.82  |      | 0.67  | 12.88 | 3.91              |                  |                               |                 |                                |
| Clinopyroxene              | 50.32            | 0.96             | 5.03                           |                                | 12.78 |      | 16.70 | 13.72 | 0.50              |                  |                               |                 |                                |
| Intermediate               | 51.25            | 1.27             | 13.93                          |                                | 11.66 | 0.17 | 7.42  | 11.37 | 2.56              | 0.09             |                               | 0.26            |                                |

## ERROR (σ) FOR OXIDES ANALYSES (%)

|                            | SiO <sub>2</sub> | TiO <sub>2</sub> | Al <sub>2</sub> O <sub>3</sub> | Fe <sub>2</sub> O <sub>3</sub> | FeO  | MnO  | MgO  | CaO  | Na <sub>2</sub> O | K <sub>2</sub> O | P <sub>2</sub> O <sub>5</sub> | SO <sub>3</sub> | Cr <sub>2</sub> O <sub>3</sub> |
|----------------------------|------------------|------------------|--------------------------------|--------------------------------|------|------|------|------|-------------------|------------------|-------------------------------|-----------------|--------------------------------|
| <b>JdF mix 2bis post</b>   |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |                                |
| Bulk rock                  | 0.05             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.02            | 0.01                           |
| Plagioclase-labradorite    | 0.08             | 0.02             | 0.05                           |                                | 0.03 |      | 0.02 | 0.03 | 0.03              | 0.01             |                               |                 |                                |
| Clinopyroxene              | 0.07             | 0.02             | 0.03                           |                                | 0.04 | 0.02 | 0.04 | 0.03 | 0.02              | 0.01             |                               | 0.02            | 0.02                           |
| Intermediate               | 0.13             | 0.04             | 0.08                           |                                | 0.07 |      | 0.05 | 0.05 | 0.05              | 0.02             |                               |                 |                                |
| Opaques                    | 0.24             | 0.34             | 0.20                           | 0.50                           |      | 0.15 | 0.20 | 0.10 |                   | 0.07             |                               |                 |                                |
| <b>CD80WP132-Abis post</b> |                  |                  |                                |                                |      |      |      |      |                   |                  |                               |                 |                                |
| Bulk rock                  | 0.04             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.02            |                                |
| Plagioclase-labradorite    | 0.11             | 0.03             | 0.08                           |                                | 0.04 |      | 0.03 | 0.05 | 0.04              |                  |                               |                 |                                |
| Clinopyroxene              | 0.26             | 0.08             | 0.12                           |                                | 0.16 |      | 0.16 | 0.12 | 0.09              |                  |                               |                 |                                |
| Intermediate               | 0.05             | 0.02             | 0.03                           |                                | 0.03 | 0.01 | 0.02 | 0.02 | 0.02              | 0.01             |                               | 0.02            |                                |

**CH4-12:** inductively coupled plasma optical emission spectrometry (ICP–OES) and ion chromatography (IC) analyses on major cations and anions in solutions, respectively. The element compositions are listed in two separated tables reporting their original values and the correspondent corrected values for the linear evaporation effect, respectively. The data are given in mmol/L. ICP-OES analytical errors are reported in a third table, and are given in percentage relative standard deviation (%RSD), which is expressed as the standard deviation of the mean as a percentage of the mean.

#### DATA – ORIGINAL VALUES (mmol/L)

| Sample      | ID     | ICP-OES |      |      |      |      |      |      | IC             |                 |      |                               |
|-------------|--------|---------|------|------|------|------|------|------|----------------|-----------------|------|-------------------------------|
|             |        | Na      | Mg   | Si   | K    | Ca   | Sr   | S    | F <sup>-</sup> | Cl <sup>-</sup> | Br   | SO <sub>4</sub> <sup>2-</sup> |
| Blank 2     | E5-1   | 484     | 53.5 | 0.02 | 10.4 | 10.9 | 0.09 | 29.3 | 0.01           | 646             | 0.78 | 33.1                          |
|             | E5-2   | 486     | 55.1 | 0.02 | 10.7 | 10.9 | 0.09 | 29.6 | 0.03           | 572             | 0.79 | 28.8                          |
|             | E5-3   | 489     | 55.1 | 0.02 | 10.6 | 11.0 | 0.09 | 29.7 | 0.05           | 591             | 0.79 | 29.6                          |
|             | E5-4   | 493     | 55.6 | 0.02 | 10.9 | 10.9 | 0.09 | 29.9 | 0.04           | 639             | 1.01 | 32.4                          |
|             | E5-5   | 496     | 56.8 | 0.02 | 10.8 | 10.9 | 0.09 | 29.9 | 0.04           | 617             | 0.76 | 30.8                          |
|             | E5-6   | 506     | 57.6 | 0.02 | 11.1 | 11.3 | 0.09 | 30.2 | 0.04           | 612             | 0.72 | 30.8                          |
|             | E5-7   | 514     | 58.0 | 0.03 | 11.5 | 11.4 | 0.10 | 30.9 | 0.01           | 616             | 1.05 | 30.7                          |
| JdF mix 1   | E6-1   | 482     | 53.9 | 0.03 | 10.3 | 10.7 | 0.09 | 29.0 | 0.02           | 622             | 0.95 | 31.2                          |
|             | E6-2   | 474     | 54.2 | 0.30 | 10.5 | 12.5 | 0.09 | 29.4 | <0.01          | 568             | 0.92 | 28.8                          |
|             | E6-3   | 483     | 53.7 | 0.40 | 10.4 | 13.1 | 0.09 | 29.8 | <0.01          | 582             | 0.96 | 29.6                          |
|             | E6-4   | 486     | 54.5 | 0.50 | 10.4 | 13.4 | 0.09 | 29.8 | 0.02           | 584             | 0.85 | 29.4                          |
|             | E6-5   | 479     | 54.2 | 0.57 | 10.3 | 13.5 | 0.09 | 29.6 | <0.01          | 569             | 0.8  | 28.8                          |
|             | E6-6   | 485     | 55.2 | 0.67 | 10.4 | 13.4 | 0.09 | 29.9 | <0.01          | 569             | 0.93 | 29.0                          |
|             | E6-7   | 480     | 54.5 | 0.77 | 10.4 | 13.6 | 0.09 | 29.7 | 0.01           | 574             | 0.91 | 29.0                          |
| JdF mix 2   | E7-1   | 478     | 53.6 | 0.04 | 10.4 | 10.9 | 0.09 | 29.2 | 0.03           | 576             | 0.78 | 29.2                          |
|             | E7-2   | 482     | 54.2 | 0.30 | 10.3 | 12.3 | 0.09 | 29.5 | 0.04           | 574             | 0.7  | 28.9                          |
|             | E7-3   | 473     | 54.1 | 0.43 | 10.3 | 12.6 | 0.09 | 29.2 | <0.01          | 568             | 0.67 | 28.6                          |
|             | E7-4   | 479     | 54.8 | 0.49 | 9.9  | 12.8 | 0.09 | 29.3 | 0.02           | 678             | 0.9  | 34.3                          |
|             | E7-5   | 480     | 55.5 | 0.57 | 10.4 | 12.9 | 0.09 | 29.4 | <0.01          | 575             | 0.69 | 28.8                          |
|             | E7-6   | 483     | 54.3 | 0.66 | 10.3 | 13.1 | 0.09 | 29.5 | <0.01          | 567             | 0.96 | 28.5                          |
|             | E7-7   | 490     | 55.8 | 0.77 | 10.5 | 13.4 | 0.10 | 30.0 | 0.01           | 585             | 0.92 | 29.5                          |
| JdF mix 3   | E8-1   | 482     | 54.6 | 0.02 | 10.4 | 10.9 | 0.09 | 29.5 | 0.04           | 581             | 0.93 | 29.5                          |
|             | E8-2   | 483     | 54.1 | 0.23 | 10.6 | 12.0 | 0.09 | 29.5 | <0.01          | 581             | 0.73 | 29.4                          |
|             | E8-3   | 485     | 55.5 | 0.38 | 10.7 | 12.2 | 0.09 | 29.5 | 0.04           | 579             | 0.81 | 29.2                          |
|             | E8-4   | 485     | 55.1 | 0.48 | 10.7 | 12.4 | 0.09 | 29.7 | 0.02           | 576             | 0.89 | 29.1                          |
|             | E8-5   | 482     | 55.2 | 0.56 | 10.6 | 12.4 | 0.09 | 29.4 | 0.02           | 687             | 1.13 | 34.5                          |
|             | E8-6   | 486     | 55.6 | 0.67 | 10.7 | 12.7 | 0.10 | 30.0 | 0.02           | 577             | 0.94 | 29.2                          |
|             | E8-7   | 486     | 55.1 | 0.80 | 10.7 | 12.5 | 0.09 | 29.7 | 0.03           | 586             | 0.8  | 29.5                          |
| Blank 3     | E9-1   | 490     | 54.8 | 0.02 | 10.2 | 11.0 | 0.09 | 29.6 | <0.01          | 579             | 0.65 | 29.4                          |
|             | E9-2   | 493     | 55.7 | 0.01 | 10.5 | 11.2 | 0.10 | 29.9 | 0.04           | 580             | 0.93 | 29.5                          |
|             | E9-3   | 494     | 55.9 | 0.02 | 10.3 | 11.0 | 0.10 | 30.1 | 0.02           | 587             | 0.72 | 29.6                          |
|             | E9-4   | 555     | 62.1 | 0.02 | 11.8 | 12.2 | 0.11 | 33.4 | 0.03           | 593             | 0.98 | 29.8                          |
|             | E9-5   | 500     | 56.4 | 0.02 | 10.5 | 11.3 | 0.09 | 29.9 | 0.05           | 593             | 0.8  | 30.1                          |
|             | E9-6   | 518     | 57.4 | 0.07 | 11.0 | 11.6 | 0.10 | 30.5 | 0.08           | 600             | 1.01 | 30.5                          |
|             | E9-7   | 527     | 58.9 | 0.02 | 10.9 | 12.0 | 0.10 | 31.5 | 0.02           | 605             | 0.82 | 30.5                          |
|             | E9-8   | 519     | 57.9 | 0.02 | 10.9 | 11.5 | 0.10 | 31.1 | 0.09           | 614             | 0.79 | 30.7                          |
|             | E9-9   | 477     | 53.4 | 0.02 | 10.3 | 10.6 | 0.09 | 28.8 | 0.02           | 621             | 0.87 | 31.3                          |
|             | E9-10  | 541     | 59.9 | 0.02 | 11.3 | 12.0 | 0.10 | 32.4 | <0.01          | 633             | 0.82 | 31.9                          |
|             | E9-11  | 546     | 61.6 | 0.03 | 11.9 | 12.4 | 0.11 | 33.3 | 0.01           | 646             | 0.93 | 32.6                          |
|             | E9-12  | 546     | 61.4 | 0.02 | 11.6 | 12.2 | 0.10 | 32.8 | 0.02           | 650             | 0.89 | 32.8                          |
|             | E9-13  | 613     | 68.5 | 0.02 | 13.1 | 13.8 | 0.12 | 36.6 | 0.08           | 666             | 1.05 | 34.0                          |
|             | E9-14  | 563     | 63.5 | 0.02 | 12.1 | 12.7 | 0.11 | 33.9 | 0.05           | 664             | 0.74 | 33.5                          |
| CD80WP132-A | E11-1  | 499     | 55.8 | 0.02 | 10.3 | 10.8 | 0.09 | 29.7 | 0.06           | 580             | 0.55 | 29.1                          |
|             | E11-2  | 494     | 56.1 | 0.14 | 10.5 | 11.1 | 0.09 | 29.8 | 0.02           | 582             | 0.66 | 29.5                          |
|             | E11-3  | 500     | 56.6 | 0.17 | 10.4 | 11.4 | 0.10 | 30.1 | <0.01          | 584             | 0.73 | 29.2                          |
|             | E11-4  | 507     | 57.1 | 0.21 | 10.7 | 11.6 | 0.10 | 30.1 | 0.08           | 591             | 0.88 | 29.6                          |
|             | E11-5  | 509     | 57.4 | 0.25 | 10.7 | 11.8 | 0.10 | 30.5 | 0.06           | 586             | 0.64 | 29.5                          |
|             | E11-6  | 500     | 56.8 | 0.27 | 10.7 | 11.4 | 0.09 | 30.0 | 0.02           | 588             | 0.79 | 29.1                          |
|             | E11-7  | 504     | 57.0 | 0.29 | 10.8 | 11.7 | 0.10 | 30.3 | 0.01           | 598             | 0.69 | 29.9                          |
|             | E11-8  | 513     | 56.8 | 0.31 | 10.8 | 11.6 | 0.10 | 30.2 | 0.15           | 599             | 0.92 | 29.9                          |
|             | E11-9  | 511     | 57.4 | 0.34 | 11.0 | 11.7 | 0.10 | 30.2 | 0.02           | 607             | 0.79 | 30.2                          |
|             | E11-10 | 513     | 57.3 | 0.37 | 11.0 | 11.6 | 0.10 | 30.6 | 0.01           | 608             | 0.74 | 30.3                          |
|             | E11-11 | 518     | 58.4 | 0.40 | 11.2 | 11.8 | 0.10 | 30.8 | 0.02           | 615             | 0.83 | 30.7                          |
|             | E11-12 | 520     | 59.2 | 0.43 | 11.2 | 12.1 | 0.10 | 31.2 | 0.04           | 626             | 0.79 | 31.5                          |
|             | E11-13 | 530     | 59.7 | 0.45 | 11.3 | 12.1 | 0.10 | 31.5 | 0.03           | 623             | 0.91 | 31.0                          |
|             | E11-14 | 522     | 59.3 | 0.45 | 11.3 | 12.2 | 0.10 | 31.3 | 0.03           | 633             | 0.93 | 31.6                          |

DATA – ORIGINAL VALUES (mmol/L), continued

| Sample         | ID     | ICP-OES |      |      |      |      |      |      | IC             |                 |                 |                               |
|----------------|--------|---------|------|------|------|------|------|------|----------------|-----------------|-----------------|-------------------------------|
|                |        | Na      | Mg   | Si   | K    | Ca   | Sr   | S    | F <sup>-</sup> | Cl <sup>-</sup> | Br <sup>-</sup> | SO <sub>4</sub> <sup>2-</sup> |
| G1-A           | E12-1  | 495     | 55.4 | 0.03 | 10.3 | 10.9 | 0.09 | 29.8 | 0.05           | 577             | 0.57            | 29.0                          |
|                | E12-2  | 493     | 56.0 | 0.16 | 10.3 | 11.8 | 0.10 | 29.8 | 0.05           | 590             | 0.82            | 29.4                          |
|                | E12-3  | 505     | 56.8 | 0.20 | 10.4 | 12.0 | 0.10 | 30.4 | 0.02           | 579             | 0.67            | 28.8                          |
|                | E12-4  | 500     | 57.3 | 0.24 | 10.4 | 12.4 | 0.10 | 30.1 | 0.02           | 587             | 0.80            | 29.5                          |
|                | E12-5  | 498     | 57.0 | 0.27 | 10.3 | 12.4 | 0.10 | 30.2 | 0.04           | 591             | 0.79            | 29.8                          |
|                | E12-6  | 493     | 56.3 | 0.30 | 10.3 | 12.2 | 0.09 | 29.3 | 0.02           | 584             | 0.73            | 29.4                          |
|                | E12-7  | 490     | 55.8 | 0.32 | 10.3 | 12.2 | 0.09 | 29.6 | 0.02           | 590             | 0.92            | 29.7                          |
|                | E12-8  | 501     | 56.2 | 0.35 | 10.3 | 12.2 | 0.09 | 29.9 | 0.01           | 595             | 0.87            | 29.6                          |
|                | E12-9  | 495     | 56.3 | 0.39 | 10.5 | 12.5 | 0.09 | 29.8 | <0.01          | 594             | 0.76            | 30.1                          |
|                | E12-10 | 507     | 56.5 | 0.41 | 10.4 | 12.4 | 0.09 | 30.1 | 0.06           | 594             | 0.77            | 30.1                          |
|                | E12-11 | 492     | 56.4 | 0.45 | 10.3 | 12.4 | 0.09 | 29.8 | <0.01          | 596             | 0.83            | 29.8                          |
|                | E12-12 | 498     | 56.6 | 0.47 | 10.4 | 12.7 | 0.10 | 30.0 | 0.01           | 599             | 0.88            | 30.1                          |
|                | E12-13 | 596     | 68.1 | 0.58 | 12.5 | 15.1 | 0.11 | 35.8 | <0.01          | 594             | 0.74            | 29.8                          |
|                | E12-14 | 499     | 57.2 | 0.50 | 10.6 | 12.7 | 0.10 | 30.3 | 0.01           | 600             | 0.84            | 30.1                          |
| Blank 4        | E13-1  | 479     | 47.8 | 0.10 | 10.6 | 10.6 | 0.09 | 29.7 | 0.13           | 539             | 1.22            | 30.6                          |
|                | E13-2  | 482     | 47.9 | 0.05 | 10.6 | 10.7 | 0.09 | 29.8 | 0.11           | 541             | 0.93            | 30.1                          |
|                | E13-3  | 476     | 47.6 | 0.06 | 10.5 | 10.6 | 0.09 | 29.8 | 0.10           | 542             | 1.22            | 30.2                          |
|                | E13-4  | 482     | 47.7 | 0.02 | 10.7 | 10.7 | 0.09 | 29.8 | 0.14           | 543             | 1.11            | 29.7                          |
|                | E13-5  | 487     | 54.0 | 0.07 | 10.4 | 11.0 | 0.09 | 29.7 | 0.15           | 543             | 0.74            | 30.5                          |
|                | E13-6  | 491     | 53.9 | 0.06 | 10.5 | 11.1 | 0.09 | 29.8 | 0.19           | 545             | 0.85            | 30.2                          |
|                | E13-7  | 490     | 54.1 | 0.06 | 10.4 | 11.1 | 0.09 | 30.0 | 0.20           | 546             | 1.45            | 31.0                          |
|                | E13-8  | 493     | 54.2 | 0.02 | 10.5 | 11.1 | 0.09 | 30.0 | 0.30           | 545             | 1.31            | 30.6                          |
| JdF mix 2bis   | E14-1  | 481     | 47.7 | 0.02 | 10.7 | 10.7 | 0.09 | 29.9 | 0.41           | 540             | 0.93            | 30.4                          |
|                | E14-2  | 483     | 48.5 | 0.16 | 10.5 | 11.0 | 0.09 | 30.0 | 0.55           | 545             | 1.14            | 30.1                          |
|                | E14-3  | 481     | 48.4 | 0.24 | 10.5 | 11.1 | 0.10 | 30.1 | <0.01          | 545             | 1.21            | 29.7                          |
|                | E14-4  | 480     | 48.7 | 0.29 | 10.6 | 11.0 | 0.10 | 30.2 | 0.13           | 543             | 1.11            | 30.3                          |
|                | E14-5  | 478     | 48.9 | 0.34 | 10.5 | 11.2 | 0.10 | 30.2 | 0.21           | 549             | 1.19            | 30.1                          |
|                | E14-6  | 493     | 54.9 | 0.42 | 10.4 | 11.4 | 0.09 | 30.0 | 0.61           | 545             | 1.10            | 30.2                          |
|                | E14-7  | 492     | 55.1 | 0.48 | 10.4 | 11.5 | 0.10 | 30.3 | 0.21           | 549             | 1.35            | 30.6                          |
|                | E14-8  | 489     | 55.4 | 0.52 | 10.3 | 11.5 | 0.09 | 30.2 | 0.27           | 552             | 1.27            | 31.4                          |
| CD80WP132-Abis | E15-1  | 475     | 47.7 | 0.02 | 10.4 | 10.5 | 0.09 | 29.6 | 0.10           | 535             | 1.42            | 29.6                          |
|                | E15-2  | 478     | 47.5 | 0.06 | 10.4 | 10.6 | 0.09 | 29.6 | 0.21           | 537             | 2.05            | 30.7                          |
|                | E15-3  | 474     | 47.4 | 0.09 | 10.4 | 10.5 | 0.09 | 29.5 | 0.18           | 532             | 0.77            | 29.9                          |
|                | E15-4  | 478     | 47.6 | 0.10 | 10.5 | 10.6 | 0.09 | 29.6 | 0.08           | 537             | 1.40            | 30.2                          |
|                | E15-5  | 479     | 50.8 | 0.13 | 10.4 | 10.7 | 0.09 | 29.7 | 0.13           | 538             | 1.08            | 30.0                          |
|                | E15-6  | 485     | 53.4 | 0.16 | 10.3 | 10.9 | 0.09 | 29.5 | 0.25           | 537             | 1.10            | 30.1                          |
|                | E15-7  | 483     | 53.5 | 0.20 | 10.3 | 10.8 | 0.09 | 29.7 | 0.08           | 539             | 1.05            | 30.3                          |
|                | E15-8  | 488     | 54.0 | 0.23 | 10.3 | 10.9 | 0.09 | 29.6 | 0.13           | 537             | 0.82            | 28.5                          |
| CD80WP132-B    | E16-1  | 479     | 48.1 | 0.01 | 10.5 | 10.8 | 0.09 | 30.1 | 0.19           | 542             | 1.15            | 30.6                          |
|                | E16-2  | 484     | 48.4 | 0.09 | 10.6 | 10.9 | 0.09 | 30.1 | 0.08           | 542             | 1.21            | 30.8                          |
|                | E16-3  | 485     | 48.2 | 0.12 | 10.6 | 10.8 | 0.09 | 30.0 | 0.35           | 546             | 1.80            | 31.0                          |
|                | E16-4  | 490     | 48.6 | 0.14 | 10.8 | 10.9 | 0.09 | 30.2 | 0.14           | 548             | 1.05            | 30.5                          |
|                | E16-5  | 494     | 54.9 | 0.17 | 10.5 | 11.1 | 0.09 | 30.1 | 0.08           | 548             | 1.19            | 31.0                          |
|                | E16-6  | 497     | 55.0 | 0.20 | 10.6 | 11.3 | 0.09 | 30.2 | 0.29           | 553             | 1.23            | 30.2                          |
|                | E16-7  | 502     | 55.1 | 0.22 | 10.7 | 11.3 | 0.09 | 30.5 | 0.19           | 554             | 0.85            | 30.4                          |
|                | E16-8  | 503     | 55.3 | 0.24 | 10.6 | 11.3 | 0.09 | 30.3 | 0.16           | 558             | 0.98            | 31.6                          |



## DATA – CORRECTED VALUES (mmol/L)

| Sample      | ID     | ICP-OES |      |      |      |      |      |      | IC             |                 |                 |                               |
|-------------|--------|---------|------|------|------|------|------|------|----------------|-----------------|-----------------|-------------------------------|
|             |        | Na      | Mg   | Si   | K    | Ca   | Sr   | S    | F <sup>-</sup> | Cl <sup>-</sup> | Br <sup>-</sup> | SO <sub>4</sub> <sup>2-</sup> |
| Blank 2     | E5-1   | 477     | 52.8 | 0.02 | 10.2 | 10.7 | 0.09 | 28.9 | 0.01           | 637             | 0.77            | 32.6                          |
|             | E5-2   | 476     | 54.0 | 0.01 | 10.5 | 10.7 | 0.09 | 29.0 | 0.03           | 560             | 0.77            | 28.2                          |
|             | E5-3   | 475     | 53.6 | 0.02 | 10.3 | 10.7 | 0.09 | 28.9 | 0.05           | 574             | 0.77            | 28.7                          |
|             | E5-4   | 475     | 53.6 | 0.02 | 10.5 | 10.5 | 0.09 | 28.8 | 0.04           | 616             | 0.98            | 31.2                          |
|             | E5-5   | 475     | 54.4 | 0.02 | 10.3 | 10.4 | 0.09 | 28.6 | 0.04           | 590             | 0.73            | 29.4                          |
|             | E5-6   | 476     | 54.2 | 0.02 | 10.4 | 10.6 | 0.09 | 28.4 | 0.04           | 576             | 0.68            | 29.0                          |
|             | E5-7   | 469     | 52.9 | 0.03 | 10.4 | 10.4 | 0.09 | 28.1 | 0.01           | 561             | 0.96            | 28.0                          |
| JdF mix 1   | E6-1   | 481     | 53.8 | 0.03 | 10.3 | 10.7 | 0.09 | 29.0 | 0.02           | 620             | 0.95            | 31.2                          |
|             | E6-2   | 473     | 54.1 | 0.30 | 10.5 | 12.5 | 0.09 | 29.3 | 0.01           | 566             | 0.92            | 28.7                          |
|             | E6-3   | 481     | 53.5 | 0.40 | 10.3 | 13.0 | 0.09 | 29.7 | 0.01           | 580             | 0.96            | 29.5                          |
|             | E6-4   | 484     | 54.3 | 0.50 | 10.4 | 13.4 | 0.09 | 29.7 | 0.02           | 581             | 0.85            | 29.3                          |
|             | E6-5   | 477     | 53.9 | 0.57 | 10.3 | 13.5 | 0.09 | 29.5 | 0.01           | 566             | 0.79            | 28.6                          |
|             | E6-6   | 481     | 54.8 | 0.66 | 10.3 | 13.3 | 0.09 | 29.6 | <0.01          | 565             | 0.92            | 28.8                          |
|             | E6-7   | 474     | 53.8 | 0.76 | 10.2 | 13.4 | 0.09 | 29.4 | 0.01           | 567             | 0.90            | 28.7                          |
| JdF mix 2   | E7-1   | 476     | 53.4 | 0.04 | 10.4 | 10.8 | 0.09 | 29.1 | 0.03           | 574             | 0.78            | 29.1                          |
|             | E7-2   | 479     | 53.8 | 0.30 | 10.3 | 12.2 | 0.09 | 29.3 | 0.04           | 570             | 0.69            | 28.7                          |
|             | E7-3   | 469     | 53.7 | 0.43 | 10.2 | 12.5 | 0.09 | 29.0 | 0.01           | 563             | 0.66            | 28.4                          |
|             | E7-4   | 474     | 54.2 | 0.49 | 9.8  | 12.6 | 0.09 | 29.0 | 0.02           | 670             | 0.89            | 33.9                          |
|             | E7-5   | 474     | 54.7 | 0.56 | 10.3 | 12.8 | 0.09 | 29.0 | 0.01           | 567             | 0.68            | 28.4                          |
|             | E7-6   | 474     | 53.3 | 0.66 | 10.2 | 12.9 | 0.09 | 28.9 | <0.01          | 556             | 0.94            | 28.0                          |
|             | E7-7   | 476     | 54.2 | 0.77 | 10.2 | 13.1 | 0.09 | 29.2 | 0.01           | 568             | 0.90            | 28.7                          |
| JdF mix 3   | E8-1   | 481     | 54.4 | 0.02 | 10.4 | 10.9 | 0.09 | 29.4 | 0.04           | 578             | 0.92            | 29.4                          |
|             | E8-2   | 480     | 53.9 | 0.23 | 10.5 | 12.0 | 0.09 | 29.4 | 0.01           | 578             | 0.73            | 29.3                          |
|             | E8-3   | 481     | 55.1 | 0.38 | 10.6 | 12.1 | 0.09 | 29.3 | 0.04           | 575             | 0.81            | 29.0                          |
|             | E8-4   | 480     | 54.6 | 0.47 | 10.6 | 12.3 | 0.09 | 29.4 | 0.02           | 570             | 0.88            | 28.8                          |
|             | E8-5   | 476     | 54.5 | 0.56 | 10.5 | 12.2 | 0.09 | 29.1 | 0.02           | 678             | 1.12            | 34.1                          |
|             | E8-6   | 478     | 54.7 | 0.67 | 10.5 | 12.5 | 0.09 | 29.5 | 0.02           | 567             | 0.93            | 28.8                          |
|             | E8-7   | 474     | 53.7 | 0.80 | 10.4 | 12.2 | 0.09 | 28.9 | 0.03           | 571             | 0.78            | 28.8                          |
| Blank 3     | E9-1   | 473     | 52.9 | 0.02 | 9.9  | 10.6 | 0.09 | 28.6 | 0.01           | 559             | 0.63            | 28.4                          |
|             | E9-2   | 473     | 53.4 | 0.01 | 10.0 | 10.8 | 0.09 | 28.7 | 0.03           | 556             | 0.89            | 28.3                          |
|             | E9-3   | 470     | 53.2 | 0.02 | 9.8  | 10.4 | 0.09 | 28.6 | 0.02           | 558             | 0.69            | 28.1                          |
|             | E9-4   | 524     | 58.6 | 0.02 | 11.1 | 11.5 | 0.10 | 31.5 | 0.02           | 560             | 0.92            | 28.1                          |
|             | E9-5   | 468     | 52.8 | 0.02 | 9.8  | 10.6 | 0.09 | 28.1 | 0.05           | 556             | 0.75            | 28.2                          |
|             | E9-6   | 481     | 53.4 | 0.07 | 10.2 | 10.8 | 0.09 | 28.4 | 0.07           | 558             | 0.94            | 28.4                          |
|             | E9-7   | 486     | 54.4 | 0.02 | 10.1 | 11.0 | 0.09 | 29.0 | 0.02           | 559             | 0.76            | 28.1                          |
|             | E9-8   | 476     | 53.1 | 0.02 | 10.0 | 10.6 | 0.09 | 28.5 | 0.09           | 563             | 0.73            | 28.1                          |
|             | E9-9   | 430     | 48.2 | 0.02 | 9.3  | 9.5  | 0.08 | 26.0 | 0.02           | 560             | 0.78            | 28.2                          |
|             | E9-10  | 481     | 53.3 | 0.02 | 10.1 | 10.7 | 0.09 | 28.9 | 0.01           | 564             | 0.73            | 28.4                          |
|             | E9-11  | 476     | 53.7 | 0.02 | 10.3 | 10.8 | 0.09 | 29.0 | 0.01           | 563             | 0.81            | 28.4                          |
|             | E9-12  | 466     | 52.5 | 0.02 | 9.9  | 10.4 | 0.09 | 28.0 | 0.02           | 555             | 0.76            | 28.0                          |
|             | E9-13  | 517     | 57.7 | 0.02 | 11.0 | 11.6 | 0.10 | 30.8 | 0.07           | 561             | 0.88            | 28.6                          |
|             | E9-14  | 471     | 53.1 | 0.02 | 10.1 | 10.6 | 0.09 | 28.3 | 0.04           | 556             | 0.62            | 28.0                          |
| CD80WP132-A | E11-1  | 490     | 54.8 | 0.02 | 10.1 | 10.6 | 0.09 | 29.1 | 0.06           | 570             | 0.54            | 28.6                          |
|             | E11-2  | 483     | 54.9 | 0.14 | 10.2 | 10.9 | 0.09 | 29.2 | 0.02           | 569             | 0.65            | 28.9                          |
|             | E11-3  | 487     | 55.1 | 0.17 | 10.1 | 11.1 | 0.09 | 29.4 | <0.01          | 568             | 0.71            | 28.5                          |
|             | E11-4  | 492     | 55.4 | 0.21 | 10.4 | 11.2 | 0.09 | 29.2 | 0.08           | 574             | 0.85            | 28.7                          |
|             | E11-5  | 492     | 55.5 | 0.24 | 10.4 | 11.4 | 0.09 | 29.4 | 0.05           | 566             | 0.62            | 28.5                          |
|             | E11-6  | 481     | 54.7 | 0.27 | 10.3 | 11.0 | 0.09 | 28.8 | 0.02           | 566             | 0.76            | 28.0                          |
|             | E11-7  | 483     | 54.6 | 0.28 | 10.4 | 11.2 | 0.09 | 29.0 | 0.01           | 574             | 0.66            | 28.7                          |
|             | E11-8  | 490     | 54.3 | 0.30 | 10.3 | 11.1 | 0.09 | 28.9 | 0.14           | 572             | 0.87            | 28.6                          |
|             | E11-9  | 483     | 54.4 | 0.33 | 10.4 | 11.0 | 0.09 | 28.5 | 0.02           | 574             | 0.75            | 28.6                          |
|             | E11-10 | 482     | 53.9 | 0.36 | 10.3 | 10.9 | 0.09 | 28.7 | 0.01           | 571             | 0.69            | 28.5                          |
|             | E11-11 | 481     | 54.2 | 0.38 | 10.4 | 11.0 | 0.09 | 28.6 | 0.02           | 571             | 0.77            | 28.5                          |
|             | E11-12 | 477     | 54.3 | 0.41 | 10.3 | 11.1 | 0.09 | 28.6 | 0.04           | 575             | 0.72            | 28.9                          |
|             | E11-13 | 483     | 54.4 | 0.41 | 10.3 | 11.1 | 0.09 | 28.7 | 0.03           | 568             | 0.83            | 28.2                          |
|             | E11-14 | 474     | 53.9 | 0.41 | 10.3 | 11.1 | 0.09 | 28.4 | 0.03           | 574             | 0.84            | 28.7                          |
| G1-A        | E12-1  | 491     | 55.0 | 0.03 | 10.3 | 10.8 | 0.09 | 29.6 | 0.05           | 573             | 0.57            | 28.8                          |
|             | E12-2  | 489     | 55.5 | 0.16 | 10.2 | 11.7 | 0.09 | 29.6 | 0.05           | 585             | 0.81            | 29.1                          |
|             | E12-3  | 500     | 56.2 | 0.20 | 10.3 | 11.9 | 0.10 | 30.1 | 0.02           | 573             | 0.66            | 28.5                          |
|             | E12-4  | 494     | 56.7 | 0.23 | 10.3 | 12.2 | 0.09 | 29.7 | 0.02           | 580             | 0.79            | 29.2                          |
|             | E12-5  | 491     | 56.3 | 0.27 | 10.2 | 12.2 | 0.09 | 29.8 | 0.04           | 583             | 0.78            | 29.4                          |
|             | E12-6  | 486     | 55.4 | 0.30 | 10.2 | 12.0 | 0.09 | 28.9 | 0.02           | 575             | 0.72            | 29.0                          |
|             | E12-7  | 482     | 54.9 | 0.32 | 10.1 | 12.0 | 0.09 | 29.1 | 0.02           | 580             | 0.91            | 29.2                          |
|             | E12-8  | 492     | 55.2 | 0.34 | 10.1 | 12.0 | 0.09 | 29.4 | 0.01           | 584             | 0.85            | 29.1                          |
|             | E12-9  | 485     | 55.1 | 0.38 | 10.3 | 12.2 | 0.09 | 29.2 | <0.01          | 581             | 0.75            | 29.4                          |
|             | E12-10 | 495     | 55.2 | 0.41 | 10.2 | 12.1 | 0.09 | 29.3 | 0.06           | 579             | 0.75            | 29.4                          |
|             | E12-11 | 478     | 54.7 | 0.44 | 10.0 | 12.0 | 0.09 | 29.0 | <0.01          | 579             | 0.81            | 28.9                          |
|             | E12-12 | 481     | 54.8 | 0.46 | 10.1 | 12.2 | 0.09 | 29.0 | 0.01           | 580             | 0.85            | 29.1                          |
|             | E12-13 | 575     | 65.6 | 0.56 | 12.1 | 14.5 | 0.11 | 34.5 | <0.01          | 573             | 0.71            | 28.7                          |
|             | E12-14 | 481     | 55.1 | 0.48 | 10.2 | 12.2 | 0.09 | 29.2 | 0.01           | 577             | 0.81            | 29.0                          |

DATA – CORRECTED VALUES (mmol/L), continued

| Sample         | ID    | ICP-OES |      |      |      |      |      |      | IC             |                 |                 |                               |
|----------------|-------|---------|------|------|------|------|------|------|----------------|-----------------|-----------------|-------------------------------|
|                |       | Na      | Mg   | Si   | K    | Ca   | Sr   | S    | F <sup>-</sup> | Cl <sup>-</sup> | Br <sup>-</sup> | SO <sub>4</sub> <sup>2-</sup> |
| Blank 4        | E13-1 | 476     | 47.5 | 0.10 | 10.6 | 10.5 | 0.09 | 29.5 | 0.13           | 535             | 1.21            | 30.4                          |
|                | E13-2 | 478     | 47.5 | 0.05 | 10.5 | 10.6 | 0.09 | 29.6 | 0.11           | 537             | 0.93            | 29.9                          |
|                | E13-3 | 471     | 47.1 | 0.06 | 10.4 | 10.5 | 0.09 | 29.5 | 0.10           | 536             | 1.21            | 29.9                          |
|                | E13-4 | 476     | 47.1 | 0.02 | 10.6 | 10.6 | 0.09 | 29.5 | 0.13           | 536             | 1.10            | 29.3                          |
|                | E13-5 | 480     | 53.2 | 0.07 | 10.2 | 10.8 | 0.09 | 29.3 | 0.15           | 535             | 0.73            | 30.1                          |
|                | E13-6 | 482     | 52.9 | 0.06 | 10.3 | 10.9 | 0.09 | 29.2 | 0.18           | 535             | 0.84            | 29.7                          |
|                | E13-7 | 479     | 52.9 | 0.06 | 10.2 | 10.8 | 0.09 | 29.3 | 0.20           | 534             | 1.42            | 30.3                          |
|                | E13-8 | 480     | 52.9 | 0.02 | 10.2 | 10.8 | 0.09 | 29.3 | 0.29           | 531             | 1.27            | 29.8                          |
| JdF mix 2bis   | E14-1 | 477     | 47.3 | 0.02 | 10.6 | 10.6 | 0.09 | 29.6 | 0.41           | 535             | 0.92            | 30.1                          |
|                | E14-2 | 478     | 48.0 | 0.16 | 10.4 | 10.9 | 0.09 | 29.6 | 0.54           | 539             | 1.13            | 29.7                          |
|                | E14-3 | 474     | 47.7 | 0.24 | 10.4 | 10.9 | 0.09 | 29.7 | <0.01          | 538             | 1.19            | 29.3                          |
|                | E14-4 | 472     | 47.9 | 0.29 | 10.4 | 10.9 | 0.09 | 29.7 | 0.13           | 534             | 1.09            | 29.8                          |
|                | E14-5 | 469     | 48.0 | 0.34 | 10.3 | 10.9 | 0.09 | 29.7 | 0.21           | 539             | 1.17            | 29.5                          |
|                | E14-6 | 481     | 53.5 | 0.42 | 10.2 | 11.2 | 0.09 | 29.3 | 0.60           | 531             | 1.07            | 29.4                          |
|                | E14-7 | 477     | 53.5 | 0.47 | 10.1 | 11.2 | 0.09 | 29.4 | 0.20           | 533             | 1.31            | 29.7                          |
|                | E14-8 | 472     | 53.5 | 0.50 | 10.0 | 11.1 | 0.09 | 29.1 | 0.26           | 533             | 1.22            | 30.3                          |
| CD80WP132-Abis | E15-1 | 473     | 47.5 | 0.02 | 10.4 | 10.5 | 0.09 | 29.5 | 0.10           | 533             | 1.41            | 29.4                          |
|                | E15-2 | 476     | 47.3 | 0.06 | 10.4 | 10.5 | 0.09 | 29.4 | 0.21           | 535             | 2.04            | 30.5                          |
|                | E15-3 | 471     | 47.1 | 0.09 | 10.3 | 10.4 | 0.09 | 29.4 | 0.18           | 529             | 0.76            | 29.7                          |
|                | E15-4 | 474     | 47.2 | 0.10 | 10.4 | 10.5 | 0.09 | 29.4 | 0.08           | 533             | 1.39            | 30.0                          |
|                | E15-5 | 475     | 50.3 | 0.13 | 10.3 | 10.6 | 0.09 | 29.5 | 0.13           | 534             | 1.07            | 29.7                          |
|                | E15-6 | 480     | 52.8 | 0.16 | 10.2 | 10.8 | 0.09 | 29.2 | 0.24           | 531             | 1.09            | 29.8                          |
|                | E15-7 | 477     | 52.8 | 0.20 | 10.2 | 10.7 | 0.09 | 29.3 | 0.08           | 532             | 1.03            | 29.9                          |
|                | E15-8 | 480     | 53.2 | 0.22 | 10.1 | 10.8 | 0.09 | 29.2 | 0.13           | 529             | 0.81            | 28.1                          |
| CD80WP132-B    | E16-1 | 474     | 47.6 | 0.01 | 10.4 | 10.6 | 0.09 | 29.8 | 0.18           | 537             | 1.14            | 30.3                          |
|                | E16-2 | 478     | 47.8 | 0.09 | 10.5 | 10.8 | 0.09 | 29.7 | 0.08           | 536             | 1.20            | 30.4                          |
|                | E16-3 | 477     | 47.4 | 0.12 | 10.5 | 10.7 | 0.09 | 29.6 | 0.35           | 537             | 1.77            | 30.6                          |
|                | E16-4 | 482     | 47.7 | 0.14 | 10.6 | 10.7 | 0.09 | 29.7 | 0.13           | 538             | 1.04            | 29.9                          |
|                | E16-5 | 483     | 53.7 | 0.16 | 10.3 | 10.9 | 0.09 | 29.5 | 0.08           | 537             | 1.17            | 30.4                          |
|                | E16-6 | 484     | 53.5 | 0.20 | 10.3 | 11.0 | 0.09 | 29.4 | 0.28           | 538             | 1.19            | 29.4                          |
|                | E16-7 | 486     | 53.4 | 0.22 | 10.4 | 10.9 | 0.09 | 29.5 | 0.18           | 537             | 0.83            | 29.4                          |
|                | E16-8 | 484     | 53.2 | 0.23 | 10.2 | 10.9 | 0.09 | 29.2 | 0.15           | 537             | 0.94            | 30.4                          |

## RSD (%) – ICP-OES

| Sample      | ID     | Na   | Mg   | Si    | K    | Ca   | Sr   | S    |
|-------------|--------|------|------|-------|------|------|------|------|
| Blank 2     | E5-1   | 0.84 | 0.51 | 12.61 | 0.40 | 0.91 | 0.31 | 0.22 |
|             | E5-2   | 1.17 | 0.27 | 7.40  | 1.59 | 0.53 | 0.44 | 0.27 |
|             | E5-3   | 1.07 | 0.18 | 13.60 | 0.66 | 0.61 | 0.25 | 0.32 |
|             | E5-4   | 0.78 | 0.79 | 9.91  | 0.83 | 0.70 | 0.21 | 0.25 |
|             | E5-5   | 0.41 | 1.13 | 2.97  | 1.72 | 1.42 | 0.26 | 0.38 |
|             | E5-6   | 0.74 | 0.75 | 10.14 | 1.14 | 0.71 | 0.62 | 0.18 |
|             | E5-7   | 0.72 | 0.73 | 5.07  | 0.94 | 1.69 | 0.31 | 0.23 |
| JdF mix 1   | E6-1   | 0.48 | 0.62 | 9.03  | 1.65 | 0.94 | 0.30 | 0.08 |
|             | E6-2   | 0.37 | 0.83 | 1.71  | 0.84 | 1.16 | 0.19 | 0.30 |
|             | E6-3   | 0.47 | 0.53 | 0.80  | 0.36 | 0.43 | 0.36 | 0.43 |
|             | E6-4   | 1.01 | 0.18 | 0.33  | 0.43 | 0.95 | 0.04 | 0.32 |
|             | E6-5   | 0.20 | 1.00 | 0.74  | 0.75 | 1.57 | 0.17 | 0.20 |
|             | E6-6   | 0.58 | 0.93 | 0.56  | 1.30 | 0.71 | 0.23 | 0.29 |
|             | E6-7   | 0.54 | 0.64 | 1.54  | 1.12 | 2.03 | 0.89 | 0.19 |
| JdF mix 2   | E7-1   | 0.65 | 0.36 | 4.91  | 0.63 | 0.74 | 0.27 | 0.38 |
|             | E7-2   | 1.33 | 1.32 | 1.74  | 1.55 | 0.88 | 0.04 | 0.14 |
|             | E7-3   | 0.97 | 0.62 | 1.23  | 0.35 | 0.58 | 0.45 | 0.07 |
|             | E7-4   | 0.95 | 1.05 | 0.25  | 8.77 | 1.72 | 0.35 | 0.13 |
|             | E7-5   | 1.23 | 0.47 | 0.47  | 1.34 | 0.70 | 0.49 | 0.08 |
|             | E7-6   | 1.35 | 1.00 | 0.75  | 0.99 | 1.02 | 0.35 | 0.04 |
|             | E7-7   | 1.54 | 0.87 | 0.88  | 1.26 | 1.06 | 0.40 | 0.34 |
| JdF mix 3   | E8-1   | 0.98 | 0.67 | 14.54 | 0.85 | 1.29 | 0.06 | 0.18 |
|             | E8-2   | 0.81 | 0.88 | 0.99  | 0.87 | 1.08 | 0.35 | 0.08 |
|             | E8-3   | 1.04 | 0.96 | 0.37  | 0.61 | 0.46 | 0.43 | 0.15 |
|             | E8-4   | 0.99 | 0.96 | 0.82  | 1.34 | 0.72 | 0.39 | 0.09 |
|             | E8-5   | 1.11 | 1.11 | 0.89  | 1.87 | 1.02 | 0.42 | 0.29 |
|             | E8-6   | 0.60 | 0.63 | 0.51  | 0.78 | 0.91 | 0.43 | 0.15 |
|             | E8-7   | 1.20 | 0.76 | 0.23  | 0.62 | 0.96 | 0.41 | 0.20 |
| Blank 3     | E9-1   | 0.27 | 0.27 | 13.85 | 0.26 | 0.77 | 0.08 | 0.07 |
|             | E9-2   | 0.28 | 0.31 | 28.68 | 0.83 | 0.10 | 0.13 | 0.32 |
|             | E9-3   | 0.74 | 0.46 | 12.41 | 3.00 | 4.98 | 0.37 | 0.35 |
|             | E9-4   | 0.51 | 0.80 | 7.03  | 0.81 | 2.46 | 0.64 | 0.21 |
|             | E9-5   | 0.71 | 0.14 | 19.23 | 1.32 | 1.25 | 0.66 | 0.36 |
|             | E9-6   | 1.52 | 1.02 | 1.71  | 1.56 | 0.39 | 0.22 | 0.15 |
|             | E9-7   | 0.64 | 0.82 | 6.83  | 1.37 | 0.34 | 0.96 | 0.07 |
|             | E9-8   | 0.95 | 0.18 | 4.74  | 1.30 | 1.31 | 0.30 | 0.17 |
|             | E9-9   | 1.66 | 0.09 | 4.70  | 0.24 | 1.12 | 0.09 | 0.18 |
|             | E9-10  | 0.12 | 0.91 | 3.68  | 0.22 | 0.58 | 0.33 | 0.02 |
|             | E9-11  | 1.21 | 0.75 | 8.68  | 0.76 | 1.25 | 0.35 | 0.27 |
|             | E9-12  | 1.72 | 0.50 | 24.88 | 0.76 | 1.12 | 0.25 | 0.18 |
|             | E9-13  | 0.69 | 0.31 | 15.80 | 0.33 | 1.48 | 0.21 | 0.28 |
|             | E9-14  | 0.76 | 0.60 | 6.77  | 0.21 | 1.04 | 0.09 | 0.59 |
| CD80WP132-A | E11-1  | 0.56 | 0.18 | 6.00  | 0.72 | 1.61 | 0.46 | 0.01 |
|             | E11-2  | 0.57 | 0.53 | 3.03  | 0.96 | 0.86 | 0.17 | 0.31 |
|             | E11-3  | 1.04 | 0.28 | 1.15  | 0.43 | 1.42 | 0.81 | 0.39 |
|             | E11-4  | 0.46 | 0.15 | 0.35  | 0.45 | 0.68 | 0.45 | 0.32 |
|             | E11-5  | 1.33 | 0.32 | 0.31  | 0.54 | 0.89 | 0.39 | 0.22 |
|             | E11-6  | 0.03 | 0.49 | 0.63  | 0.68 | 0.68 | 0.29 | 0.28 |
|             | E11-7  | 0.90 | 0.50 | 0.39  | 0.94 | 1.97 | 0.13 | 0.14 |
|             | E11-8  | 0.76 | 0.45 | 1.42  | 2.02 | 0.47 | 0.60 | 0.32 |
|             | E11-9  | 0.26 | 0.47 | 0.35  | 0.79 | 0.39 | 0.21 | 0.77 |
|             | E11-10 | 0.81 | 0.47 | 1.37  | 0.67 | 0.36 | 0.57 | 0.28 |
|             | E11-11 | 0.67 | 0.36 | 0.99  | 0.90 | 0.65 | 0.42 | 0.08 |
|             | E11-12 | 1.11 | 0.34 | 0.86  | 0.62 | 0.68 | 0.25 | 0.19 |
|             | E11-13 | 0.18 | 0.23 | 0.64  | 0.47 | 0.26 | 0.16 | 0.61 |
|             | E11-14 | 1.08 | 0.71 | 0.57  | 1.29 | 0.87 | 0.38 | 0.01 |
| G1-A        | E12-1  | 0.99 | 0.19 | 6.34  | 0.38 | 1.15 | 1.02 | 0.24 |
|             | E12-2  | 0.29 | 0.23 | 1.81  | 1.57 | 0.43 | 0.53 | 0.95 |
|             | E12-3  | 0.22 | 0.44 | 0.29  | 0.30 | 0.63 | 0.53 | 0.02 |
|             | E12-4  | 0.90 | 0.58 | 0.27  | 0.54 | 0.51 | 0.28 | 0.52 |
|             | E12-5  | 0.40 | 0.24 | 0.76  | 0.54 | 0.85 | 0.47 | 0.12 |
|             | E12-6  | 0.56 | 0.52 | 0.31  | 0.91 | 0.95 | 0.35 | 1.07 |
|             | E12-7  | 0.51 | 0.90 | 0.73  | 0.62 | 0.63 | 0.34 | 0.16 |
|             | E12-8  | 0.95 | 0.70 | 0.83  | 0.64 | 1.39 | 0.22 | 0.35 |
|             | E12-9  | 0.81 | 0.14 | 0.61  | 1.07 | 0.79 | 0.18 | 0.52 |
|             | E12-10 | 0.90 | 0.16 | 0.29  | 1.53 | 1.45 | 0.72 | 0.32 |
|             | E12-11 | 0.86 | 0.57 | 0.73  | 0.94 | 0.97 | 0.51 | 0.27 |
|             | E12-12 | 0.57 | 0.50 | 0.42  | 0.93 | 1.58 | 0.43 | 0.32 |
|             | E12-13 | 1.31 | 0.25 | 0.70  | 0.96 | 1.32 | 0.35 | 0.26 |
|             | E12-14 | 0.57 | 0.27 | 0.88  | 1.28 | 1.05 | 0.17 | 0.24 |

RSD (%) – ICP-OES, continued

| Sample         | ID    | Na   | Mg   | Si    | K    | Ca   | Sr   | S    |
|----------------|-------|------|------|-------|------|------|------|------|
| Blank 4        | E13-1 | 0.94 | 0.53 | 2.50  | 0.71 | 1.08 | 0.62 | 0.21 |
|                | E13-2 | 0.66 | 0.60 | 4.66  | 1.02 | 0.84 | 0.40 | 0.31 |
|                | E13-3 | 0.56 | 0.42 | 2.99  | 1.09 | 0.82 | 0.44 | 0.31 |
|                | E13-4 | 0.81 | 0.51 | 5.02  | 0.75 | 0.87 | 0.38 | 0.19 |
|                | E13-5 | 0.69 | 0.47 | 5.58  | 0.74 | 1.00 | 0.25 | 0.25 |
|                | E13-6 | 1.05 | 0.60 | 3.87  | 1.26 | 1.02 | 0.20 | 0.23 |
|                | E13-7 | 1.09 | 0.32 | 3.99  | 0.76 | 0.83 | 0.59 | 0.31 |
|                | E13-8 | 1.00 | 0.36 | 8.56  | 1.21 | 0.87 | 0.43 | 0.25 |
| JdF mix 2bis   | E14-1 | 0.87 | 0.51 | 10.56 | 0.94 | 0.72 | 0.36 | 0.31 |
|                | E14-2 | 0.40 | 0.56 | 1.28  | 1.06 | 0.79 | 0.44 | 0.33 |
|                | E14-3 | 1.01 | 0.27 | 0.54  | 0.72 | 0.71 | 0.36 | 0.25 |
|                | E14-4 | 0.62 | 0.26 | 0.73  | 0.88 | 0.88 | 0.46 | 0.29 |
|                | E14-5 | 0.73 | 0.24 | 0.86  | 1.09 | 1.03 | 0.45 | 0.17 |
|                | E14-6 | 0.70 | 0.50 | 0.50  | 1.04 | 0.80 | 0.46 | 0.16 |
|                | E14-7 | 0.58 | 0.29 | 0.60  | 0.56 | 0.77 | 0.40 | 0.19 |
|                | E14-8 | 1.17 | 0.51 | 0.76  | 0.93 | 0.85 | 0.33 | 0.27 |
| CD80WP132-Abis | E15-1 | 0.60 | 0.25 | 9.73  | 1.13 | 0.63 | 0.45 | 0.28 |
|                | E15-2 | 0.70 | 0.41 | 2.71  | 1.00 | 0.73 | 0.30 | 0.25 |
|                | E15-3 | 0.85 | 0.59 | 2.72  | 1.13 | 0.74 | 0.32 | 0.13 |
|                | E15-4 | 0.78 | 0.29 | 1.20  | 0.54 | 0.68 | 0.32 | 0.21 |
|                | E15-5 | 0.78 | 0.51 | 1.48  | 1.62 | 0.99 | 0.28 | 0.32 |
|                | E15-6 | 0.73 | 0.44 | 0.73  | 0.78 | 1.03 | 0.38 | 0.22 |
|                | E15-7 | 1.01 | 0.29 | 0.84  | 1.51 | 1.87 | 0.59 | 0.26 |
|                | E15-8 | 0.58 | 0.63 | 0.94  | 0.62 | 0.90 | 0.31 | 0.18 |
| CD80WP132-B    | E16-1 | 0.68 | 0.42 | 14.70 | 0.92 | 0.69 | 0.54 | 0.24 |
|                | E16-2 | 0.50 | 0.66 | 1.82  | 0.72 | 0.98 | 0.35 | 0.18 |
|                | E16-3 | 0.78 | 0.54 | 2.06  | 0.83 | 1.32 | 0.49 | 0.26 |
|                | E16-4 | 0.44 | 0.71 | 0.99  | 1.09 | 0.96 | 0.42 | 0.16 |
|                | E16-5 | 0.69 | 0.31 | 1.25  | 0.91 | 0.99 | 0.39 | 0.16 |
|                | E16-6 | 1.28 | 0.43 | 0.84  | 1.00 | 0.98 | 0.49 | 0.39 |
|                | E16-7 | 0.50 | 0.31 | 1.07  | 0.80 | 1.01 | 0.34 | 0.28 |
|                | E16-8 | 0.95 | 0.46 | 1.25  | 0.99 | 0.67 | 0.34 | 0.24 |

## A-Chapter 4, Samples

**CH4-13:** Complete list of samples, divided by textural group, from the scientific cruise IODP 327, at Site U1362, Hole A.

### JdF mix 1, cryptocrystalline basalts

15 samples



**JdF mix 2, microcrystalline basalts**

14 samples



**JdF mix 3, altered fine grained basalts**  
11 samples





**CH4-14:** Pillow basalt from Mid-Atlantic Ridge (MAR) dredged during the *RRS Charles Darwin Cruise 80*, specifically at the *Way Point 132*.

**CD80WP132, cryptocrystalline vesicular basalt**



**CH4-15:** Gabbro from Troodos ophiolite (Cyprus), sampled during a fieldwork close to the locality of Kato Amiandos.

**G1, recrystallized medium grained gabbro**





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