Chemical Society Reviews



Chem Soc Rev

A review of Mineral Carbonation Technologies to sequester CO2

Journal:	Chemical Society Reviews
Manuscript ID:	CS-REV-01-2014-000035.R1
Article Type:	Review Article
Date Submitted by the Author:	30-May-2014
Complete List of Authors:	Sanna, Aimaro; Heriot-Watt University, Engineering and Physical Sciences Uibu, Mai; Tallinn University of Technology, Laboratory of Inorganic Materials Caramanna, Giorgio; Heriot-Watt University, Engineering and Physical Sciences Kuusik, Rein; Tallinn University of Technology, Laboratory of Inorganic Materials Maroto-Valer, Mercedes; Heriot Watt University, Engineering and Physical Sciences
	1



A review of Mineral Carbonation Technologies to sequester $\ensuremath{\text{CO}_2}$

A. Sanna^a*, M. Uibu^b, G. Caramanna^a, R. Kuusik^b, M. M. Maroto-Valer^{a,c}

^a Centre for Innovation in Carbon Capture and Storage (CICCS), School of Engineering and Physical Sciences, Heriot-Watt University, Edinburgh EH14 4AS, UK

^b Laboratory of Inorganic Materials, Tallinn University of Technology, 5 Ehitajate S., Tallinn 19086, Estonia

^c Institute of Petroleum Engineering, Heriot-Watt University, Edinburgh EH14 4AS, UK

*Corresponding author. Address: School of Engineering and Physical Sciences, 3.04 Nasmyth Building, Heriot-Watt University, Edinburgh EH14 4AS, UK. Tel.: +44(0)131 451 3299. E-mail address: A.Sanna@hw.ac.uk

Abstract

Carbon dioxide (CO₂) capture and sequestration includes a portfolio of technologies that can potentially sequester billions of tonnes of CO_2 per year. Mineral carbonation (MC) is emerging as a potential CCS technology solution to sequester CO₂ from smaller/medium emitters, where geological sequestration is not a viable option. In MC processes, CO₂ is chemically reacted with calcium- and/or magnesium-containing materials to form stable carbonates. This work investigates the current advancement of the proposed MC technologies and the role they can play in decreasing the overall cost of this CO₂ sequestration route. In-situ mineral carbonation is a very promising option in terms of resources available and enhanced security, but the technology is still in its infancy and transport and storage costs are still higher than geological storage in sedimentary basins (\$17 instead of \$8/tCO₂). Ex-situ mineral carbonation has been demonstrated in pilot and demonstration scale. However, its application is currently limited by its high costs, which range from \$50 to $$300/tCO_2$ sequestered. Energy use, reaction rate and materials handling are the key factors hindering the success of this technology. The value of the products seems central to render MC economically viable in the same way as conventional CCS seems profitable only when combined with EOR. Large scale projects such as the Skyonic process can help in reducing the knowledge gaps on MC fundamentals and provide accurate costing and data on processes integration and comparison. The literature up-to date indicate that in the coming decades MC can play an important role in decarbonising the power and industrial sector.

1. Introduction

1.1 Carbon Capture and Storage

Our knowledge of the global carbon cycle is sufficiently extensive to conclude that natural processes cannot absorb all the anthropogenically produced carbon dioxide (CO₂) in the coming centuries, so that adaptation technologies are urgently required ¹. Extensive evidence on the anthropogenic cause of climate change can be obtained by the comprehensive IPCC report published in 2007 ². There is a general agreement that to meet the ambitious target to stabilize atmospheric CO₂ concentration at 500 ppm by 2050, a large portfolio of technologies need to be considered, where Carbon Dioxide Capture and Storage (CCS) represents a leading technology, particularly in the transition from a fossil fuel based economy to a renewable based economy ³.

CCS refers to a number of technologies which capture CO_2 at some stage from processes such as combustion for power generation, cement manufacture, iron and steel making and natural gas

treatment. Then, the captured CO_2 is pressurised (≥ 100 bar) prior to being transported (by pipeline, ship, rail or road) to a storage site, where it is injected into stable geological sites (saline aquifers, depleted oil and gas fields, deep coal seams), trapping it for thousands of years ^{3,4}.

 CO_2 capture at power plants and other large point sources represent the most likely tool towards the reduction of current CO_2 emissions from fossil fuels use. CCS is not a new concept and a large number of different CO_2 capture technologies are being developed, ranging from currently commercial technologies such as amine-scrubbing with geological storage through 2nd or 3rd generation technologies, such as chemical or carbonate looping. Recent improvements in amine scrubbing and processes have reduced the energy requirement from 450 kWh/ tCO₂ (2001) to about 200-300 kWh/t in 2012, which will result in a reduction of the power output by 20 to 30% in a typical coal-fuelled power plant (1000 kwh/tCO₂ output)^{4,5}.

Even if most of the individual components of the CCS chain (e.g. capture) have been demonstrated, their integration into a single process is challenging and has still to be demonstrated ^{4,6}.

Moreover, the delay of some large demonstration projects (e.g. Mongstad in Norway) due to higher complexity than expected and general public acceptance issues related to potential leakages and surface transport of supercritical CO_2 are delaying the deployment of geological storage ⁴. Overall CO_2 geological storage poses a great deal of uncertainty in terms of quantification of storage potential, monitoring of injected CO_2 and engineering challenges to ensure that the injected CO_2 remains in the subsurface for hundreds or thousands of years ⁴.

Under this scenario, Mineral Carbonation (MC) represents an alternative CCS option, which may be particularly suitable for small sources.

1.2 CO₂ sequestration by Mineral Carbonation

Mineral carbonation (MC) is an accelerated form of weathering of natural occurring silicate rocks and has been proposed as an alternative approach for CO_2 sequestration since the 1990s⁷. Some MC technologies are recently approaching the commercial stage. MC is defined as the reaction of metal oxide bearing materials with CO_2 to form insoluble carbonates:

Metal oxide + $CO_2 \rightarrow$ Metal-carbonate + Heat (1)

This reaction can take place either below (in-situ) or above (ex-situ) ground. In-situ mineral carbonation involves the injection of CO_2 into underground reservoirs to promote the reaction between CO_2 and alkaline-minerals present in the geological formation to form carbonates⁸. Ex-situ mineral carbonation relates to above-ground processes, which requires rocks mining and materials comminution as MC pre-requisite⁹.

MC can take advantage of different starting materials, which include Mg-silicate minerals and Ca⁻ or Fe⁻-rich silicates. The reactions occurring in MC processes are listed below ⁹.

$Mg_2SiO_4 + 2CO_2 + 2H_2O \rightarrow 2MgCO_3 + H_4SiO_4$	(2)
$Mg_{3}Si_{2}O_{5}(OH)_{4} + 3CO_{2} + 2H_{2}O \rightarrow 3MgCO_{3} + 2H_{4}SiO_{4}$	(3)
$Fe_2SiO_4 + 2CO_2 + 2H_2O \rightarrow 2FeCO_3 + H_4SiO_4$	(4)
$CaSiO_3 + CO_2 + 2H_2O \rightarrow CaCO_3 + H_4SiO_4$	(5)

CCS by geological storage represents the best CCS strategy for large emitters, while MC can play an important role by targeting small and medium emitters (< 2.5 Mt CO₂), which account for about 10-15% of the total CO₂ emissions ¹⁰. Mineral carbonation is a permanent and safe way for storing CO₂, which does not present potential concerns over long term monitoring and liability issues, such as geological storage. The inherent stability of mineral carbonation is confirmed by the distribution of carbon in the lithosphere of the Earth (Figure 1), where approximately half of the total carbon is in the form of limestone (CaCO₃) and other types of carbonates.

Mineral carbonation resources have a large CO_2 sequestration potential (>10,000 Gt C) due to the large abundance of silicates around the world, as shown in Figure 2. MC could also be sourced by CO_2

extracted directly from the air or the ocean. A method to extract CO_2 from ambient air has been proposed adopting components and fabrication methods derived from cooling towers and using strong NaOH solution. The cost of this process (without taking into consideration the regeneration of NaOH) was of $60/tCO_2$ ¹¹. However, other works have quantified the costs of this option to be as high as 600-1000 per tCO_2 ¹². Moreover, Goldberg et al. estimated that ~75 Mt CO_2/y could be collected using air capture powered by wind energy and sequestered below seafloor in basalts formations at Kerguelen (see also Section 2.2), where regional reservoirs could hold over 1500 Gt CO_2 , sequestering a large fraction of 21^{st} century emissions ¹³. In addition, CO_2 has been extracted from seawater. A total of 59% of dissolved inorganic CO_2 in seawater has been extracted using bipolar membrane electrodialysis with an energy input of ~1527 kWh/tCO₂¹⁴. While the extraction of CO_2 from air and seawater has been demonstrated, many challenges remain, including slow extraction rates, poor CO_2 selectivity and high costs.

One of the issues associated to MC is the disposal/use of the products. Despite the fact that only a fraction of the MC products would be absorbed by current market for carbonates and silica as shown in Figure 1, the disposal capacity considering mine and land reclamation projects around the world is considered feasible since large reclamation projects can involve 100-200Gt of materials ¹⁰. Moreover, MC products (carbonates) may be used for ocean liming ^{15,16}. It has been calculated that MC–Ocean Liming would require approximately 4.9 and 2.2 GJ of thermal and electrical energy ton⁻¹ of CO₂ sequestered ¹⁶. A lab-scale seawater/mineral carbonate gas scrubber was found to remove up to 97% of CO₂ in a simulated flue gas stream at ambient temperature and pressure, with a large fraction of this carbon ultimately converted to dissolved calcium bicarbonate ¹⁵. However, manipulation of ocean chemistry may create additional environmental impact on marine life, such as localized elevated pH or co-dissolution of trace metals ¹⁶.

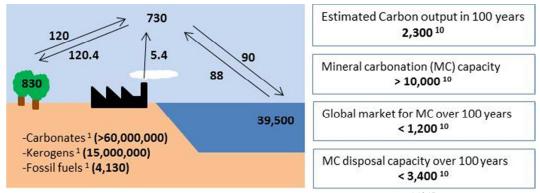


Figure 1. Global carbon reservoirs (GtC) and net fluxes (GtC/y) (modified from ^{1,10,17}).

Mineral resources availability, scalability, applicability to regions without geologic storage capacity, inherent stability of the reactions products and the potential revenue from MC products is supporting the on-going development of this technology. Also, mineral carbonation can operate on flue gases directly, without CO_2 pre-separation, which typically stands for 70-75% of the costs of the CCS chain ^{18,19}. It has been reported a CO_2 carbonation efficiency of ~20% when SO_x and NO_x were present in the flue gas (15% CO_2) using wollastonite at 40bar and 150°C ²⁰. However, very few works have been published on MC in presence of impurities to fully assess this option.

MC effectiveness for CO_2 mitigation purposes has been limited due to the slow kinetics of the CO_2 silicate reactions, energy intensive pre-treatments, logistic issues (e.g. locations of mineral resources and CO_2 emitters, development of transport and storage facilities for waste carbonates on a large scale) and scalability issues (e.g. a coal-fired power station fitted with 100% mineral sequestration would require more tonnes of mineral feedstock than of coal)^{10,17}. Several reviews on mineral carbonation technologies focused on a number of processes under development have recently been produced ^{8,17,21,22}. Olajire ²¹ focused his review on in-situ technologies and the environmental impact of the reaction products with regard to their possible beneficial utilization. Zevenhoven et al ²² reviewed the state-of-the-art of ex-situ mineralisation, illustrating the future prospects of CO₂ mineralization and including a portfolio of CCS technologies under development worldwide. Salek and co-workers ¹⁷ reviewed the potential sequestration of CO₂ using environmental biotechnological processes, such as nitrification, anaerobic digestion (AD) and bio-electrochemical systems. The global sequestration potential of biodegradable solid waste and wastewater by AD employing silicates was estimated in 0.2-0.4 wt.% of the total anthropogenic CO₂ emissions. Bioelectrochemical systems such as microbial fuel cells can potentially sequester more CO_2 than what is produced during the organic carbon oxidation (200 wt%). However, these systems suffer from extremely low current densities and therefore further development is required ¹⁷. Although these bio-based processes are promising, their development is still in its infancy and is not considered in this review. Also, a number of reviews on mineral carbonations of alkaline wastes have been recently published ^{23,24}, providing an overview of the types of industrials wastes that can be used for mineral carbon sequestration and the process routes available. They concluded that industrial waste residues represent an alternative source of mineral alkalinity and are readily and cheaply available close to CO₂ sources, and moreover, the carbonation of waste residues often improves their environmental stability.

The objective of this paper is to provide an up-to date review and discussion on mineral carbonation technologies, including potential deployment for in-situ and ex-situ carbonation in presence of silicate rocks and alkaline wastes and products applications. Technologies using mineral rocks as feedstock are treated in more depth, since previous work clearly indicate that alkaline wastes represent a niche application and would only be able to marginally reduce the global CO_2 emissions (<1%)^{25,26}.

In-situ mineral carbonation is analysed in the first section of the work, followed by the evaluation of the silicate rocks for ex-situ mineral carbonation technologies based on minerals and inorganic wastes feedstocks. Finally, a section on the potential products from mineralisation and final remarks are presented.

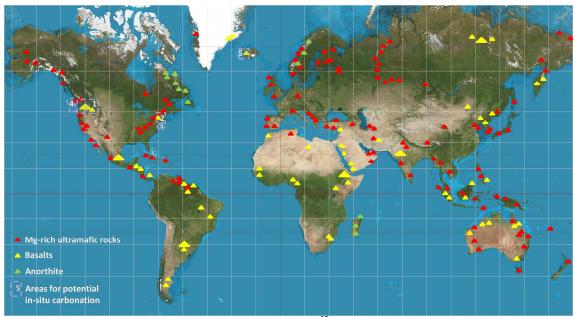


Figure 2. Mineral feedstock available for MC. (modified from ²⁷).

2. In-situ mineral carbonation

Carbonation is a natural process where CO_2 reacts with different minerals forming solid precipitates leading to the weathering of the rocks. The reactions are spontaneous and exothermic and can be exemplified as (6) and (7) where calcium and magnesium oxides are considered reacting with CO_2 .

$$CaO + CO_2 = CaCO_3 + 179 \text{ KJ/mol}$$
(6)

$$MgO + CO_2 = MgCO_3 + 118 \text{ KJ/mol}$$
(7)

The most reactive compounds for CO_2 mineralization are oxides of divalent metals, Ca and Mg and their availability in nature is mainly in the form of silicates, such as olivine ((Mg,Fe)₂SiO₄) orthopyroxene (Mg₂Si₂O₆ – Fe₂Si₂O₆), clinopyroxene (CaMgSi₂O₆ – CaFeSi₂O₆) and serpentine ((Mg, Fe)₃Si₂O₅(OH)₄), the latter originated by the hydratation of olivine. When CO₂ dissolves in water, it reacts with these silicates forming corresponding carbonates, where CO₂ is fixed in a mineral form ^{28,29}.

Mantle peridotite and basalts deposits both, enriched in Mg, Fe and Ca silicates are the main targets for in-situ CO_2 mineralization projects, as discussed below²¹.

2.1 Peridotites

Peridotite is a component of ophiolites which are complex geological sequences representing the emplacement on land of sections of oceanic crust ³⁰. The world's largest ophiolitic outcrop is the Samail Ophiolite in the Sultanate of Oman extending for about 350 km with a width around 40 km and average thickness of 5 km; about 30 vol% is composed by mantle peridotite. The mineralogical composition of this peridotite is 74% olivine (partially serpentinized), 24% orthopyroxene, 2% spinel (MgAl₂O₄) and traces of clinopyroxene ^{31,32}. The Samail Ophiolite is characterized by the presence of an internal network of fractures hosting aquifers of variable volumes and chemical compositions, where several mineralized springs emit alkaline waters enriched in carbonates; the origin of those waters is linked to the natural carbonation process acting within the peridotite (Figure 3).

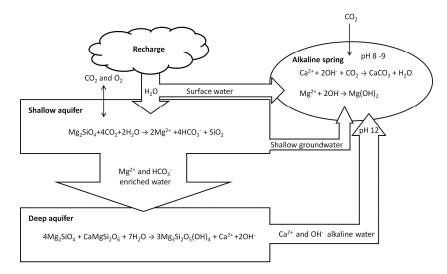


Figure 3. Water fluxes and chemical reactions within peridotites.

Surface water flows through fractures originating a shallow aquifer open to the atmospheric CO₂ and O_2 fluxes; the water reacts with the peridotite and the pre-existing carbonate rocks in an open system becoming enriched in Mg^{2+} and HCO^{3-} . This water may infiltrate in the deeper regional aquifer which is isolated from the atmospheric fluxes. The chemical reactions with the peridotite will trigger the formation of serpentine, brucite, magnesite and dolomite; Ca²⁺ and OH⁻ will accumulate in the water leading to a strong pH increase up to a value of 12. When these waters emerge at the surface in the alkaline springs the sudden intake of CO₂ from the atmosphere will precipitate Ca-carbonates; the mixing with the shallow aquifer will further precipitate Ca-carbonates and brucite. The formation of carbonates, mostly in the form of terraced travertine around the springs, consumes some OH decreasing the pH to values of 8-9. An estimate of the total volume of carbonate in the Samail Ophiolite is of 5.5 $\pm 10^7$ m³ with an average age of 26,000 years indicating that about 4 ± 10^7 Kg CO₂/year are consumed by the precipitation of carbonates. This natural process requires long times for the reactions to develop, in the order of magnitude of 50 years for the shallow-water aquifers and up to 5,600 years for the deep reservoirs. Artificial enhancement of the carbonation can be achieved injecting fluids with a higher concentration of CO₂ and increasing the temperature. For example, when injecting CO₂ at 90 °C with 100 bar pCO₂, about 0.63 kg of CO₂ can be permanently stored as carbonates for each kg of peridotite. A typical in-situ mineralization project in the Samail Ophiolite could include the drilling of the peridotite, hydrofracturing of the hosting volume, injection of heated fluids to increase the temperature at 185 °C, which is the optimum temperature for olivine carbonation rates, followed by injection of pure CO₂ at 25 °C. The exothermal reaction (producing 760kJ/kg) and the geothermal gradient (up to 20 °C/km) will both contribute to the reduction in the energy needed for heating the fluids. The resulting enhancement of the carbonation rate following this process is considered to be one million times faster than the natural process pace ^{32,33}.

2.2 Basalts

The largest presence of basalts is on the oceanic crust ³⁴. Large outcrops of basalts are also present on the continental crust ³⁵. Basalts can have a good degree of secondary permeability due to the formation of altered and brecciate horizons or networks of fractures during or after their deposition. The resulting pore space may be filled by circulating waters originating aquifers within the hosting rocks at different depth and mineral concentration. These aquifers are often enriched in ions including Ca²⁺ and Mg^{2+ 36}, which can react with the injected CO₂ precipitating carbonates and releasing H⁺ as in reaction (8):

$$(Ca^{2+}, Mg^{2+}) + CO_2 + H_2O = (Ca, Mg)CO_3 + 2H^+$$
 (8)

The reaction rate is controlled by the concentration of H^+ and will not proceed further until these ions are neutralized by the reaction with the hosting rock. Considering olivine and Ca-plagioclase basalts the neutralization process follows the reactions (9) and (10).

$Mg_2SiO_4 + 4H^* = 2Mg^{2*} + SiO_2$	(9)
$CaAl_2Si_2O_8 + 8H^+ = Ca^{2+} + 2Al^{3+} + 2SiO_2(aq) + 4H_2O$	(10)

The availability of reactive Mg, Al and Ca silicates is therefore the controlling factor for the development of in-situ CO_2 mineralization ³⁷.

Following the injection of CO_2 (both as supercritical fluids or as aqueous solution) the dissolution of some minerals and the precipitation of others, mostly carbonates, may change the porosity of the reservoir; carbonates deposition during the first stages of the injection may have adverse effects on the storage potential due to the reduction in available pore space which is progressively filled by minerals, and thus, clogging the surrounding of the injection well. Mineral deposition in a more advanced phase of the injection and during the post-injection phase instead is considered an advantage enhancing the trapping potential of the hosting structure.

Injecting CO_2 within the basalts of the ocean seafloor would benefit from a further series of trapping mechanisms in addition to the geochemical transformation of CO_2 in carbonates. The deep water environment, below 2,700 m, and the cold temperature, below 2°C, will make the injected CO_2

denser than the surrounding seawater, allowing it to sink with a gravitation-trapping mechanism; the same environmental parameters are also favourable to the formation of CO_2 hydrates, where the CO_2 molecule is "encaged" within a lattice of ice strongly reducing its solubility in water in case of seepage. Lastly, the thick sedimentary cover of the seafloor will form a low-permeability layer further reducing the possibility of leakage³⁸.

Currently a few injection-test projects and feasibility studies are addressing the potential of basalts, both onshore and offshore, for CO₂ storage (Tab. 1; Fig. 2).

Table. 1. Location and characteristics of in-situ basaltic carbonation areas.

Location	Reservoir	Caprock	Storage potential	References	
New York state	Palisides sill.	Lacustrine	An injection test	37 - 39	
(USA)	Dolerite with Ca- plagioclase and pyroxenes. Target zone porosity 5%.	deposits of the Newark Basin; mudstone, arkoses, carbonate nodules shale and clastic	was aimed to identify the buffering potential of dolerite.		
Washington, Oregon and Idaho states (USA)	Columbia River Basalt Group. Clinopyroxene, plagioclase and glass. Over 300 overlapping flows. Total volume in excess of 224,000 Km ³ . Target zone porosity 15 to 25%	sequences. Basalts layers with very low permeability.	10 to 50Gt CO ₂	40-41	
Offshore USA East Coast	Sandy Hook Basin basalt Porosity 15%	Sedimentary cover, mudstone, silt, clay	900Mt of CO ₂	52	
Offshore USA West Coast	Juan de Fuca Plate basalts. Pillow lava and massive flows. Average porosity 10%	Fine-grained turbiditic sequences and clay deposits	920 Gt of CO ₂	48	
Iceland	Ultrabasic to basic (45 – 49% SiO2) basalt flows and hyaloclastite of olivine-tholeiite composition.	Low-porosity basalts	12 MtCO ₂	43-45	

2.2.1 Onshore basalts

To validate the geochemical reactions during CO_2 storage in basalts, small-scale tests were conducted in eastern US using a single-well push-pull experiment at the contact zone between the 230 m thick Palisides sill, composed by dolerite with Ca-plagioclase (CaAl₂Si₂O₈) and pyroxenes, and the underlying sedimentary rocks of the Newark Basin composed by lacustrine deposits; mostly mudstone, arkoses, carbonate nodules shale and clastic sequences. The estimated porosity of the target zone is 5%. A fluid solution enriched in NaCl and KBr as tracers and equilibrated with CO_2 at 8 bar and pH 3.5 was injected for 3 hours in a 8 m thick section of dolerite; the total injected volume was 1.4 m³ and was followed by a further 208 litres of tap water used to flush the solution into the formation surrounding the well. After 7 days the pull phase was started to collect samples. Within hours the pH returned to the initial value of 8.3; dissolution of plagioclase and pyroxene within the groundwater was measured as consequences of the fluids injection. From these results it was estimated that 1,000 m³ of CO₂ saturated water at P_{CO2} 8 bar will be neutralized within 19 hours after the injection. Dissolution/precipitation processes and influence on the porosity of the reservoir are not totally understood yet ^{37,39}. The Columbia River Basalt Group is a large continental flood basalt deposit covering an area of more than 200,000 km² in Washington, Oregon and Idaho states in the North-West US. Originated by a series of fissural eruptions between 17.5 and 6.0 million years ago, it is composed by over 300 different flows with a volume in excess of 224,000 km³. The flows can be divided in top and bottom areas, showing a vesicular and scoriaceous crust often brecciated, and an inner massive core. The mineral composition is clinopyroxene, plagioclase and glass, the latter ranging from a few percent to over 50%. The basalts host a regional deep aquifer of brackish water with high levels of sulphide and fluoride exceeding the limits for drinking water; this aquifer is therefore the target for CO_2 injection at a depth between 663 m and 887 m thorough a well drilled in the Walla Walla County in southeastern Washington State. The hosting rocks are part of a top layer area with permeability in the range of 75 to 150 md with 15 to 25% of porosity calculated from uncalibrated sonic logs. The water chemistry of the injection area is of sodium-bicarbonate type with pH 9.68, fluoride 4.98 mg/L and Fe 962 mg/L. In July 2013 a test injection of 1,000 t CO₂ was started for one month. A model simulation forecasts that after one year 18% of the CO₂ will be dissolved in groundwater within a radius of 55 m from the bore-hole. The storage capacity of the total basalt deposits for CO₂ in situ mineralization is estimated from 10 to 50Gt $CO_2^{40,41}$.

Iceland has large volumes of basalt flows and hyaloclastites (volcanic breccias originated by the contact between the emitted lava and water or ice) associated with strong volcanic and geothermal activity with release of large volumes of CO_2 . There are evidences of natural carbonation processes within the aquifers hosted in the basaltic deposits. The Hekla volcano in Iceland originated from linear eruptions during the last 900 years and represents the deposition of basaltic andesitic tephra largely composed by volcanic glass. The water feeding a series of springs has high alkalinity and pH ranging from 7.7 to 9.3; the variation in DIC (Dissolved Inorganic Carbon) within the spring water and its correlation with changes in pH is considered a proof that CO_2 is fixed as carbonates. Another relevant aspect is that the heavy metals ions which can be mobilized from the rocks by the acidification induced by the CO_2 fluxes are incorporate in carbonates and oxy-hydroxides once the pH increases thus reducing the risk of environmental pollution ⁴².

In Iceland the CarbFix project aims to assess the feasibility of in-situ carbonation in basalt using as CO2 source the gas emitted from the geothermal power plant of Hellisheidi. The gas is associated with the geothermal steam and is composed of 83% CO_2 , 16% H_2S and the remaining 1% as H_2 , N_2 , CH_4 ; the gas stream is condensed separating CO_2 and H_2S , which is mostly re-injected in the deep geothermal reservoir. The resulting final gas is composed by 98% of CO_2 plus 2% of H_2S and it is to be injected in the deep basaltic aquifer. The hosting rocks are ultrabasic to basic (45 - 49% SiO₂) basalt flows and hyaloclastites (a breccia rich in volcanic glass) of olivine-tholeiite composition. Crystalline lava flows were in place after the last glacial age and the hyaloclastites originated during the last glaciations under the ice cover in a time span between 116,000 and 1,500 years ago. The resulting structure is a sequence of more permeable lava flows hosting a shallow aquifer underlined by lowpermeability layers of hyaloclastites separating and isolating a deeper aquifer. The shallow aquifer (above 400 m) has temperature of 8 – 12 °C, pH 7.7 – 8.4, CO_2 in balance with the atmospheric values, it is enriched in O_2 and it is undersaturated in calcite. The groundwater of the deep aquifer has temperature between 18 and 33 °C, pH of 8.4 – 9.4, CO₂ concentration below the atmospheric balance and it is depleted in O₂ and saturated in calcite. This deep aquifer between 400 and 800 m is the target area for the storage in a volume of about 1 km³; CO₂ will be dissolved in water at P_{CO2} of 25 bar, pH 3.7, DIC 1 mol/kg, requiring 22 t of water for each ton of CO2. A total of 2,200 tCO2/y will be injected. A monitoring program including geochemical analysis, tracers and isotopic concentration measures will assess the diffusion of the injected CO₂ within the aquifer and the changes in chemistry associated with the mineralisation reaction. Models show that calcite precipitation will reduce the porosity of the reservoir of about 1%. The total storage potential is of about 12 MtCO₂, or 200 years considering the annual emission from the geothermal plant in 60,000 tCO $_2$ ⁴³⁻⁴⁵.

In 2011-2012 a test injection of 175 tons of pure CO_2 and water was performed and in 2012-2013, 130 tons of CO_2 mixed with H_2S from the power plant were injected in a low-temperature (20-50 °C)

aquifer at 400-800 m of depth and in deeper reservoir (below 800 m) at higher temperature (>250 $^{\circ}$ C) 46 .

2.2.2 Offshore basalts

In order to quantify the suitability of CO_2 storage in submarine basalts some areas considered of large storage potential have been studied; their main characteristics are good porosity, presence of sealing deposits at the top and confined aquifers hosted within the formation where CO_2 can dissolve. An example of such areas is the Juan de Fuca Plate, offshore western USA, which is characterized by basaltic bedrock of high average porosity (10-15%) composed by pillow lavas and massive flows covered by fine-grained turbiditic sequences and clay sediments ⁴⁷. The pillow lavas and the altered and fractured areas have the highest porosity, up to 20%; the massive flows show a much lower porosity ranging between 2 and 9%. Hydrothermal fluids circulate within the bedrock and are effectively contained by the sealing effect of the overlying deposits allowing for long waterrock interaction times and permitting chemical reaction with the surrounding altered basalt. The target for CO_2 injection is a volume of 7,800 km³ of altered basalt with an average porosity of 10% for a storage potential of 920 Gt of CO_2^{48} .

Another large accumulation of basalts is present along and offshore the eastern coast of North America where it was originated by a series of floods during subsequent eruptive events resulting in strata separated by alteration horizons with vesicular and brecciated layers ⁴⁹. A series of deep sediments-filled basins have been identified offshore and, both from geophysical data and cores, basalt layers have been identified below the sedimentary cover ^{50,51}. These offshore basalts could represent a large storage volume for the CO₂ produced by point sources in the highly industrialized eastern shore belt of the USA.

As example, the basalt present as bedrock of the Sandy Hook basin offshore New Jersey has an average porosity of 15% with an available pore volume for CO_2 storage of about 1 km³ able to host 900Mt of CO_2 which trapping will be enhanced by long-term mineral carbonation reactions⁵².

A general estimation of the overall potential storage capacity for ocean basalts is of 8,238 Gt of CO₂; this value is calculated considering CO₂ storage in brines hosted in 20 m thick horizons of relatively porous (10% average porosity) pillow lavas and flows at depth below 2,700 m and overlaid by at least 200 m of ocean sediments. The water depth and sedimentary cover enhance the gravitational, hydrate formation and low-permeability sealing trapping mechanisms ⁵³.

3. Ex-situ carbonation

Most processes under consideration for mineral carbonation focus on metal oxide (such as calcium and magnesium) bearing materials, whose corresponding carbonates are not soluble. Moreover, since waste materials rich in calcium oxide are conveniently located close to the CO_2 emission source, they have also been targeted as MC feedstock. The following sections will review the processes developed for both rocks and wastes resources.

3.1. Processes developed for minerals

Since oxides and hydroxydes of Ca and Mg are not abundant, silicates rocks containing the desired Mg and Ca have been targeted for mineral carbonation ⁹. Table 2 summarises the main minerals available and their performance in terms of mass ratio of ore necessary to carbonate the unit mass of CO₂ (R_{real}) and reaction efficiency (E_{CO2}). Serpentine, olivine, and to less extent wollastonite because of its lower abundance, are preferred based on performance and availability ⁵⁴.

The sequestration of CO_2 in carbonates can be achieved through various process routes, which are described in this section: 1) Direct carbonation (DC) is the simplest approach, where a Ca/Mg rich solid is carbonated in a single process step. DC can be further divided into gas-solid carbonation and direct aqueous mineral carbonation. The direct aqueous mineral carbonation-route with the aid of

pre-treatments (DCP) is considered as the state of the art and is typically selected to compare other technologies. 2) Indirect carbonation (IC) consists of first extracting from the feedstock the reactive Mg/Ca oxide or hydroxide in one step and then, in a subsequent step, reacting the leached cations with CO_2 to form the desired carbonate⁸.

Table 2. Mineral chemistry, carbonation potential, and reactivity (modified from ⁵⁴). (Carbonation test conditions: 80% -37 μ m feed; 1 hour; 185°C; PCO2=150 atm; 15% solids; 0.64 M NaHCO, 1 M NaCl). R_{real} = Mass ratio of ore necessary to carbonate unit mass of CO₂; E_{CO2}% = Reaction efficiency, % stoichiometric conversion of Ca, Fe²⁺, and Mg cations in silicate feed to carbonate.

Rock	Mineral	Mg	Са	Fe ²⁺	R _{real}	Е _{со2} , %
Serpentine	Antigorite	24.6	< 0.1	2.4	2.1	92
Serpentine	Lizardite	20.7	0.3	1.5	2.5	40
Olivine	Fayalite	0.3	0.6	44.3	2.8	66
Olivine	Forsterite	27.9	0.1	6.1	1.8	81
Feldspar	Anorthite	4.8	10.3	3.1	4.4	9
Pyroxene	Augite	6.9	15.6	9.6	2.7	33
Basalt		4.3	6.7	6.7	4.9	15
Oxide	Magnetite	0.3	0.6	21.9	5.5	8
Ultramafic	Talc	15.7	2.2	9.2	2.8	15
Ultramafic	Wollastonite	0.3	31.6	0.5	2.8	82

For each process carbonation route, key parameters of a process are presented in Table 3 as to give quick insight into the main features. The parameters considered are the feedstock material used; the mineral to CO_2 ratio (R_{real}); the mineral pre-treatment type (M-mechanical, C-chemical, T-thermal); operational temperature and pressure; chemical additives used and finally, the mineral cation reacted with CO_2 (degree of conversion) (E_{CO2} %). The different direct and in-direct processes are discussed in details in Section 3.1.2.

Table 3. Main characteristics and mineral carbonation of minerals ($^{18-19,54,68,70,72,75-80,82,91,95-96,99-101,107-108,109-114,120-123,125}$).

Resource	R _{real}	Process conditions	ECO ₂ , %	Remarks
Serpentine	3.39	DCP: M, T (75µm, 630°C),115 bar, 155°C, 0.64 M NaHCO ₃ , 1M NaCl, 6 hours ^{54,82}	62%	cost of 210 \$/tCO ₂ stored (lizardite)
Serpentine	5.25	DCP: M,T (75µm:630°C), 115 bar, 155°C, 0.64 M NaHCO ₃ ,1M NaCl, 1 hour ^{95,96}	80%	cost of about A\$22/t CO ₂
Serpentine	n.a.	DCP: M,T serp. (650°C), 30-45 bar, 140°C, 150-200µm, NaHCO3 ^{18,19,91}	n.a.	flue gas is used
Serpentine	3.33	IC: M (37μm), 120 bar, 300°C, NH ₃ , 1 hour, ¹²³	70%	
Serpentine	2.33	DCP: M (100µm), 50 bar, 550°C, Weak acid/NH ₃ , 1 hour ¹²⁵	90%	
Serpentine	3	IC: M (75μm),1 bar, 140°C (300°C regeneration), 1.4M NH ₄ HSO ₄ , NH ₃ , 2-3 hours ^{68,72}	70%	
Serpentine	5.38	IC: Μ (100μm), 40 bar, 65°C, H2SO4, NaOH, 3 hours, ¹⁰⁹	55%	
Serpentine	5.83	IC: M,T (250-425µm), 20 bar, 500-550°C, (NH ₄) ₂ SO ₄ , NH ₃ , 0.5 hour ¹²⁰⁻¹²²	60%	Only 50-60% Mg is recovered
Serpentine	8.97	IC: M,T (50µm:150), 1 bar, 325°C, HCl/NaOH, 26 hours ¹⁰⁸	26%	
Serpentine	3.5	IC: M (<75µm), 20 bar, 120°C, 0.1M citrate, EDTA, 2 hours ^{110,111}	60%	
Serpentine	3.23	IC: M (75µm), 1 bar, 70°C, 1 vol% orthophosphoric acid, 0.9 wt% oxalic acid, 0.1% EDTA, 1 hour 113,114	65%	Internal agitation with grinding media is used in the dissolution stage costs between 50 and 100 \$/tCO ₂
Olivine	2.22	DCP: M (38µm), 150 bar, 185°C, 0.64M NaHCO ₃ , 1M NaCl, 3 hours, ^{54,82}	68%	stored
Olivine	1.82	DCP: M (<200µm), 30 bar, 140°C, NaHCO3, 1-2 hours ^{18,19}	80-90%	
Olivine	2.8	DC: M, C (10µm), 1.2 bar, 500°C, Bases, 0.5 hour, ^{76,77}	25%	A Spray Dry Absorber (SDA) is used
Olivine	n.a.	DC: M (2.5-60μm), 500°C, 0.5 hour, moisture ⁷⁵	12%	flue gas is used dissolution of olivine in carbonated aqueous solutions using a flow-
Olivine	n.a.	IC: M (75μm), 250°C, 150 bar, 2 hours ¹⁰⁷	11%	through column reactor
Olivine	n.a.	DC: M (37µm), 97 bar, 80°C, 20 hour ^{78,79}	8%	
Nollastonite	n.d.	DCP: M (38µm), 40 bar, 100°C, distilled water, 1 hour ^{54,82}	80%	
9	4.06	DC: M (38µm), 20 bar, 200°C, 0.25 hour ^{70,80}	75%	cost of 102 euro/ton CO_2 stored
Nollastonit				90% dissolution was achieved in 1
2	n.a.	IC: 80°C, 30 bar, succinic acid ¹¹²	n.a.	hour, but carbonation was not tested
Brines	n.a.	DCP: 1 bar, 100°C, seawater/brine/alkalinity (NaOH or Mg(OH) $_2$ or MgO) 99,100	70%	
				amounts of NaOH and /or electricity.
Brines	n.a.	DCP: 30°C, NaOH, NaCl, Electrolysis ¹⁰¹ treatment; T=thermal pre-treatment; n.a.= nov available; DC = direct carbonation; D	98%	Potential co-removal of SOx, NO2.

= indirect carbonation

3.1.1 Process chemistry and reaction kinetics

The main barriers to the commercial deployment of carbonation are high energy intensity, low reaction conversion and slow reaction kinetics ^{8,9}. To reach the highest carbonation efficiency the controlling mechanisms and optimal parameters need to be defined.

The solid particle dissolution process is generally controlled by: 1) diffusion through a fluid film surrounding the particle, 2) diffusion through a solid product layer on the particle surface, or 3) chemical reaction at the particle surface ⁵⁵. The rate of the overall process is controlled by the slowest of these sequential steps. Dissolution kinetics for olivine and serpentine, the two main source silicate minerals for mineral carbonation, have been studied for several decades; especially, olivine has attracted noticeable interest ⁵⁶⁻⁶⁰.

Dissolution of mineral is the rate-limiting step in direct aqueous mineral carbonation system, mainly due to absence of protons at pH close to 7 ⁶¹⁻⁶³. In aqueous–solid reaction system, the rate-limiting step is the dissolution of the mineral followed by product layer diffusion control (i.e. silica layer reduces diffusion of CO₂ or carbonate precipitate). In CO₂-water-solid systems, the reaction rate of CO₂ dissolution (gas diffusion through fluid film control) is the limiting-control step ⁶³. Despite the fact that dissolution rates of minerals are commonly believed to be proportional to their crystals surface, so that the precipitate; a recent aqueous dissolution tests on diopside (25-70°C, in presence of NaHCO₃) did not show any difference between the dissolution rate of the experiments with and without carbonate precipitation ⁶⁴. It has been suggested that the precipitation of carbonate forms a porous coating on the mineral, which allows ions from the dissolving mineral to be transported readily to the bulk fluid.

Particle size is a key parameter affecting carbonation because the reduction of size increases surface area and consequently the availability of reactive Mg and Ca⁶⁵. The deposition of an inert layer such as SiO₂ on the surface limits the diffusion of the extraction solution into the particles⁶⁶. Stirring or sonication during carbonation limits the formation of carbonate shells, allowing further dissolution of Mg and Ca and diffusion of CO_2^{65} . Temperature influences both the dissolution of CO_2 in water (dissolution decreases with temperature) and dissolution of calcium and magnesium from the minerals (dissolution increases when temperature rises). Low temperatures enhance the diffusion of CO_2 in the carbonated shell, while high temperatures increase the magnesium and calcium available ^{65,67}. For example, temperatures of 90-100°C are able to extract 100% of magnesium from serpentine mineral ⁶⁸.

High pressure (40-150 bar) can be used to enhance both the dissolution of CO_2 in the water media and the diffusion of the gas into the solid matrix ⁶⁹. CO_2 pressure and stirring rate (1500rpm) can significantly influence the reaction rate in direct aqueous carbonation at the optimum temperature range (150-200°C). Operating above those values, carbonation was considered independent of the stirring rate and CO_2 pressure. The carbonation of wollastonite at constant temperature (150°C) remains constant between 10 and 40 bar and decreases at CO_2 pressure below 10 bar due to deficiency of (bi)carbonate activity. On the other hand, the wollastonite carbonation increases when the CO_2 pressure is increased from 20 to 40 bar at 200-225°C, due to deficiency of (bi)carbonate activity ⁷⁰.

The control mechanisms of carbonation of pure CO_2 and flue gas carbonation are expected to be similar, but reaction rate of diluted CO_2 is slower because its dissolution rate is slower compared to that of pure CO_2 ⁷¹. Liquid/solid (L/S) ratio is an important parameter because carbonation requires specific L/S ratios to be efficient ⁶⁵. L/S-ratios lower than 2 cannot be stirred sufficiently in an autoclave reactor and may result in poor CO_2 gas-liquid and solid-liquid mass transfer rates. Therefore, the lowest liquid-to-solid ratio in an autoclave reactor is 2 kg/kg, although the majority of aqueous carbonation experiments are carried out at higher L/S-ratio to enhance the conversion efficiency ⁷². Also, a reduction of the L/S-ratio leads to a substantial improvement of the heat balance of the process and, thus, the overall CO_2 sequestration efficiency. However, if the L/S-ratio becomes too low, pumping and stirring problems might arise because of an increased viscosity, which would lead to a significant decrease of the conversion. Theoretically, an industrial process might be operated to 1:1 L/S ratios improving the overall CO_2 sequestration efficiency of the process ⁷⁰.

3.1.2 Direct Carbonation (DC)

DC consists of (1) gas-solid carbonation and (2) aqueous carbonation.

3.1.2.1 Direct Carbonation without pre-treatments

Gas-solid carbonation: The most straightforward process route is the direct gas-solid carbonation ⁷³, and it was first studied by Lackner and co-workers ⁷⁴. Various reactions depending on the feedstock are possible. As an example, the direct gas-solid reaction of olivine is given:

$$Mg_2SiO_4(s) + 2CO_2(g) = 2MgCO_3(s) + SiO_2(s)$$
 (11)

High CO₂ pressures (100-150bar) are necessary in order to obtain reasonable reaction rates. DaCosta and co-workers ⁷⁵ developed a direct dry process for the sequestration of CO₂ where the flue gases pass through a bed of finely ground (2.5-60 μ m) silicate rocks (mainly olivine, serpentine or wollastonite). As carbonation takes place, the mineral is replenished by either feeding fresh mineral with a pump or a conveyer. They reported that when using 5g of olivine (surface of 2.5m²/g) at temperature ranging from 100 to 500°C and flue composition of 10% CO₂, 8.3% H₂O (balanced with N_2), the storage capacity was 0.12g CO₂/g olivine (12%) after 30 minutes ⁷⁵. A higher capacity of 18% was achieved capturing flue gas with 15% CO₂ in presence of 8.3% water at 150°C. The CO₂ stored decreased when 5% or 20% CO₂ gas stream were used, in absence of moisture and at the higher temperatures tested, 175 and 200°C ⁷⁶⁻⁷⁷. The enhanced CO₂ stored capacity in presence of moisture was related to the fact that water vapour can be useful to convert oxides that may be present to hydroxides which may then be carbonated as in equations 12 and 13:

$MgO+H_2O\rightarrow Mg(OH)_2$	(12)
Mg(OH)₂+CO₂→MgCO₃	(13)

The above process is able to work in a dry environment, where moisture present in flue gas is assumed to be enough to convert the silicates oxides in the high reactive hydroxides and also requires only ~10-30 minutes, which represent a time scalable to industrial level. However, it would require a large amount of mineral per tonne of CO_2 sequestered due to the low efficiency (< 20 g CO_2/g olivine)⁷⁶. Based on the data available, more than 8 tonnes of olivine would sequester 1 tonne of CO_2 . This would drastically reduce the applicability of this process to very small CO_2 emitters in terms of process size and materials handling. Also, particle size reduction to <60 µm is very energy intensive.

Aqueous carbonation: The carbonic acid route process involves CO_2 reacting at high pressure (100-159 bar) in an aqueous suspension with olivine or serpentine ^{61,69}. Firstly, CO_2 dissolves in the water and dissociates to bicarbonate and H⁺ resulting in a pH of about 5.0 to 5.5 at high CO_2 pressure:

$$CO_2(g) + H_2O(I) = H_2CO_3(aq) = H^+(aq) + HCO_3(aq)$$
 (14)

 Mg^{2+} is then liberated from the mineral matrix by H^+ :

$$Mg_{2}SiO_{4}(s) + 4H^{+}(aq) = 2Mg^{2+}(aq) + SiO_{2}(s) + 2H_{2}O(l)$$
(15)

Finally, the Mg²⁺ reacts with bicarbonate and precipitates as magnesite:

$$Mg^{2+}(aq) + HCO^{3-}(aq) = MgCO_3(s) + H^{+}(aq)$$
 (16)

Kwak and co-workers ^{78,79} investigated the reaction pathways and reaction extent of direct aqueous carbonation of finely ground olivine (forsterite) (1g) mixed with water (1g) and fed into a batch reactor with a volume of 11.7mL. The reaction was kept at 80°C and 97 bar for 20 h with a final CO_2 storage capacity of 8%. The capacity was increased to 67% but it requred 7 days.

Huijgen and co-workers ⁸⁰ studied the direct aqueous carbonation of finely ground wollastonite mineral to particle size 38µm that was suspended in distilled water. A CO₂ stream was introduced into the reactor under continuous stirring to ensure dispersion of the gas. The carbonation reactions occur in the aqueous phase in two steps: calcium leaching from the CaSiO₃ matrix and nucleation and growth of CaCO₃. A promising conversion of 75% was attained after 15 minutes at 200°C, 20 bar CO₂ partial pressure, with estimated costs of 102 \notin /t CO₂ sequestered, based on process simulation (Aspen). The major costs were associated with the feedstock and the electricity consumption for grinding and compression, with 54 and 26 \notin /t CO₂ sequestered, respectively ⁷⁰.

Overall, direct routes present straightforward design and absence of non-aqueous solvents. However, reaction conversions are low and high CO_2 pressure and temperature are required, compared to processes where pre-treatments are used to enhance the CO_2 storage capacity ⁷³. To enhance to reaction conversion, various pre-treatments have been employed.

3.1.2.2 Direct Carbonation with pre-treatment

The purpose of the pre-treatment step is to promote and accelerate carbonation reaction rates and efficiencies through surface area increase. Two major processes have been developed: high energy mechanical grinding and chemical leaching, although other methods such as thermal- and mechano-chemical- pretreatments have also been reported.

3.1.2.2.1 Direct Carbonation with mechanical pre-treatment

The mechanical grinding approach aims at destroying or disordering the mineral lattice, and thus, resulting in an increase of the surface area. Particle size reduction takes place in a sequence of crushing and grinding stages required to reduce the particle size to < $300 \,\mu$ m which can be necessary to liberate valuable mineral grains. Crushing is normally performed on dry materials using compression equipment such as jaw or cone crushers. Instead, grinding is accomplished by abrasion and impact of the ore by the free motion of unconnected grinding media such as rods, balls, or pebbles⁸¹.

The US National Energy Technology Laboratory (NETL) developed a direct carbonation process (scheme shown in Figure 4), involving grinding of magnesium (or calcium) silicates at 150-200 °C, 100-150 bar, where 0.64 M NaHCO₃ and 1 M NaCl were added to the solutions ^{54,82}. NaHCO₃ was used to turn to slightly alkaline pH the solution in order to facilitate carbonate precipitation. Olivine carbonation proceeded to over 80% in 6 h. Wollastonite was found to be the most reactive, reaching over 70% in 1 h, and unlike the magnesium minerals, the wollastonite reaction proceeded rapidly in distilled water ⁸². The carbonation of olivine and wollastonite was controlled by the surface area consistent with the shrinking-core model, in which the particle surface reacts to release magnesium into solution, leaving a shrinking core. The higher wollastonite efficiency was related to the much higher precipitation rate for CaCO₃ compared to MgCO₃, which is four orders of magnitude lower than those of CaCO₃⁸³.

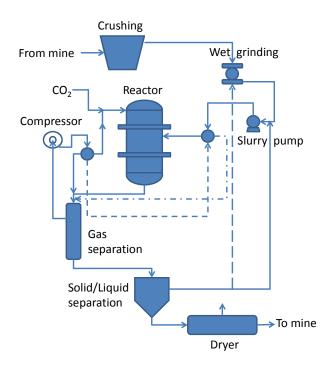


Figure 4. Scheme of NETL process (modified from ⁸²).

Various pre-treatment options such as ultrasonic treatment and wet grinding in caustic solution have also been tested, but they did not result in a higher reactivity ⁶¹. The major problem with many other pre-treatment options is the high energy input required ⁸⁴. Extensive studies in mechanical activation

of silicates were performed at NETL ^{85,86} and were reviewed later by Huijgen & Comans ⁸⁴ and Zevenhoven and co-workers ⁸. The major conclusions made were that high-energy attrition grinding of silicates resulted in a higher conversion rates but consumed too much energy.

Similar conclusions were obtained by Fabian *et al.*⁸⁷ who studied the CO₂ storage capacity of olivine mechanically activated using different conditions and by Haug et al.⁵⁷ who reported dissolution (in 0.1M HCl, and pressured CO₂/H₂O) and carbonation (115-128°C and under 150-185 bar) rates of grinded olivine. Activation in a planetary mill, even if effective, was found to consume too much energy (see Table 4) for CO₂ sequestration purposes. Therefore, it can be concluded that activation methods, such as thermal and chemical activation (discussed in the following sections), are preferred options to mechanical activation.

Table 4. Energy consumption for mineral feedstock pre-treatments^{82,87,88}. (306) energy consumption including mechanical pre-treatment.

Mechanical activation (olivine)	RPM	Time, min.	SA, m²/g	Crystalline phase,%	Energy consumption kWh/t
As-received	-	-	0.25	100	-
		10	7.3	45	190
Attritor	1500	30	18.1	26	580
		120	35.2	12	2310
Planetary	450	10	4.8	37	2010
Fidiletary	430	30	5.2	17	6030
Nutating	900	10	3	57	640
Attritor	1500	10	7.3	n.a.	170
Attritor	1500	30	18.1	n.a.	520
Attritor	1500	120	35.2	n.a.	2080
Thermal activation (serpentine)			Carbonation efficiency %		Energy consumption kWh/t
Antigorite (75µm)			62		293 (306)
Antigorite (38µm)			92		293 (376)
Lizardite (75µm)			9		326 (339)
Lizardite (38µm)			40		326 (409)

3.1.2.2.2 Direct Carbonation with thermal-pre-treatment

As previously mentioned, serpentine requires additional thermal-treatment to remove hydroxyl groups, resulting in the chemical transformation to pseudo-forsterite. Serpentine requires heating treatment above 630°C to remove chemically bound water from the lattice ⁸⁹.

 $Mg_3Si_2O_5(OH)_4 \rightarrow (MgO)_3(SiO_2)_2 + 2 H_2O$

(17)

The NETL findings indicate that the reaction rate for serpentine was slow if water (OH groups) was not removed. Thermal-treated serpentine at 630°C for 2 hours reached 65% CO₂ storage capacity. Similar results were obtained with high-energy attrition grinding, but with a substantial associated energy penalty ^{54,82}. The theoretical energy required for the heat-activation process is the sum of the energy to heat the mineral to 630°C and the enthalpy of dehydroxylation. As shown in Table 4, this energy (as electrical power) was quantified in 293 and 326 kWh/t for antigorite and lizardite, respectively ⁵⁴. Other authors have performed thermal-treatment optimisation studies ⁹⁰⁻⁹⁴. Sanna et al. ⁹⁴ reported that the energy requirement for 0.5 h activation at 610°C could be lowered to 245 kW h/t instead of 326 kW h/t previously reported (630°C for 2 h). This enhanced the subsequent dissolution of serpentine from 60% to 90% in just 5 minutes, where the Mg extracted was higher compared to another recent work where thermal activation was performed at 640–700°C for 1 h ⁹².

The direct use of thermal heat instead of electrical energy, coupled to partial dehydroxylation with heat integration (63% decrease in energy requirement for thermal-activation) has led to an overall mineral carbonation process estimated cost of A\$ 70/ tCO_2 avoided ^{91,93}, compared to \$ 210 / tCO_2 avoided in the NETL process ⁸². Balucan et al. ⁹³, studied the thermal activation of serpentine from the Great Serpentinite Belt in New South Wales (Australia), in a Thermo-Gravimetric Analysis-Differential Scanning Calorimetry (TGA-DSC) apparatus. This serpentine was found to be particularly suitable for heat activation to a 20% residual hydroxyl groups, as opposed to the partly serpentinised ultramafic minerals of the Coolac Serpentinite Belt. The activation strategy comprised heating serpentine particles with diameter < 34µm to 680°C (1.5 h) to produce an active material with 20% residual hydroxyl groups and the recovery of ~80% of the sensible heat from the dehydroxylated mineral. This results in a thermal activation estimated cost of A\$ 1.25/t of serpentine ⁹³. Serpentine from Coolac Serpentinite Belt used in the experiments had much larger particle size (D_{90} 127 μ m) so that the comparison is rather difficult. Also, particles size reduction to $< 34 \mu m$ is energy intensive (>220 kWh/t)⁸². However, if these preliminary studies are confirmed ⁹¹, technologies that use thermal-pretreatment may become more attractive, although they may be constrained by the specific reactivity of the serpentine used. Overall, on the data available, thermo-treatment is more effective in accelerating the Mg extraction than mechanical activation, although its associated energy penalty still remains significantly high ^{87,94}.

3.1.2.2.3 NETL derived processes

Brent and co-workers explored a better use of the system heat in order to avoid the drawbacks of serpentine thermal-chemical activation ⁹⁵. The process is being exploited by Orica, a large Australian company interested in mineral carbonation. The energy savings were obtained after preheating the mineral feedstock in a combined series of heat exchangers utilising the exothermic heat from the carbonation reactor and low grade heat from the same power plant that provides the flue gas. The proposed mineral carbonation plant does not include CO₂ capture and serpentine mining. Serpentine was reduced to a particle size of less than 75 μ m. Since the recovered low grade heat from carbonation (120-150°C) and power plant was not enough to reach the desired activation temperature (> 580°C), a hydrocarbonaceous fuelled furnace was used for the last heating step. Same chemical additives as in the NETL process were used. Carbonation was carried out at pressure in excess of 150 bar. This process has been claimed to be economically viable (Aspen modelling) for the permanent storage of 14.1 Mt/y of CO_2 emissions from a conventional pulverised fuel electricity generation plant (15,500 GWh per annum), which would consume about 41 Mt/y of serpentine and an additional 0.9 Mt/y coal to activate the serpentine at a claimed cost of about A22/t CO₂ ^{95,96}. However, for the process neither experimental nor simulation work have been reported so far and it presents a very high grade of process integration, which may be difficult to achieve. Moreover, it does not consider some logistic issues, such as long distance location of mineral and CO₂ point source.

Shell has developed an aqueous slurry-based mineralisation technology suitable for both serpentine and olivine mineral rocks. The process comprises pre-treatment, leaching and precipitation steps, where activation of serpentine is achieved by both mechanical and thermal means ^{18,19}. The overall process resembles that developed by NETL, but operates at lower process conditions. The slurry from the leaching step is pressurised (up to 45 bar) and heated up to 110-140°C in the precipitation step. Here, precipitation of dissolved Mg(HCO₃)₂ takes place as well as transformation of hydromagnesite into magnesite. Two different concepts have been proposed as shown in Figure 5, namely Shell's direct pure and flue gas mineralization technology (Figure 5, a - b). Since flue gas with 10%vol CO₂ has a much lower solubility than pure CO₂ under pressure, leaching of cations in presence of CO₂ will take place at a much slower rate. To avoid this, the flue gas is contacted with the mineral slurry previous precipitation stage in a separate slurry mill at ambient temperature ¹⁹. The slurry mill achieves both a huge reduction of particle size and the formation of carbonate intermediates other than bicarbonate, for instance hydromagnesite. Shell's thermal activation, which can reduce energy requirement up to 63%, has been employed in this process ⁹¹. It consists of heating the serpentine (preferably 150-200µm) for 1 hour at temperature of 650°C in fluidised bed. No data on the CO₂ sequestration capacity and energy consumption are publically available. Technical feasibility of Shell's proposed direct flue gas mineralization concept using activated serpentine has been proved at the Energy Centre of the Netherlands (ECN) in a continuous pilot plant. The continuous experiments show that seawater accelerates rate of leaching and subsequent precipitation, but requires particular material choice as expected. Also, it was shown that yielding dissolved magnesium bicarbonate by not using a magnesium carbonate precipitation unit has a large cost advantage, which was quantified in about 80% cost reduction. On the basis of the information available, the proposed process seems to be able to decrease the energy requirements for the serpentine pre-treatment, compared to the NETL process by employing proprietary thermo-treatment technology, but no information on potential costs are available yet. This technology would not get any benefit from the resultant products, since they are mixed in slurry, which would require energy intensive processing to isolate them.

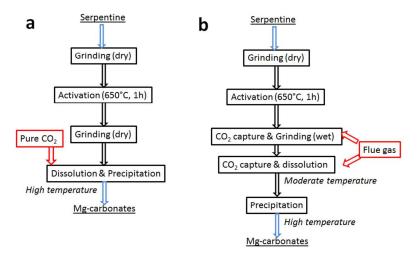


Figure 5. Mineralization process concept for pure CO₂ and flue gas (modified from ¹⁹).

3.1.2.2.4 Brines based processes

A different approach has been developed by Calera, which owns a demonstration plant at the gas fired Moss Landing power plant (USA). The Moss Landing plant has demonstrated to capture flue gas CO_2 from a 10MW power generator at 90% efficiency for about 2 years ^{96,97}. A diagram of the Calera process is depicted in Figure 6. The technology foresees the introduction of brines containing alkaline earth metal ions to a reactor, where the brine is contacted with CO_2 containing gas. The CO_2 dissolves in water to produce carbonate and bicarbonate ions, resulting in a decrease in the pH of the solution. In order to produce carbonate – containing precipitate, protons are removed from the solution so as to shift the equilibrium towards carbonate (which requires pH 9-11 to precipitate). The solution pH is then raised through the introduction of alkalinity to the point where the alkali metals are precipitated as carbonates, which may be suitable for cement manufacture. A 20% replacement in blended cement appeared technically feasible, but not yet demonstrated ⁹⁸. An energy penalty ranging from about 10% to 40%, depending on power plant characteristics and availability of brines has been associated to this process ⁹⁸.

Besides the fact that a large number of brines containing alkaline earth metal ions (Ca, Mg) are mentioned as potential feedstocks ^{99,100}, the technical suitability of brines (e.g. initial proposal of using brines at the Latrobe Valley demonstration project was abandoned because of technically

unsuitable brines), sea water and alkaline waste for the Calera process (design at the Moss Landing California pilot plant concluded that sea water required too much energy and alkaline industrial waste would be too limited for sustainable operations at significant scale) reduce a wide application of this technology ⁹⁸.

Figure 6 provides a scheme of the Calera process, where the final product is a cementitious material with the consistency of mud, which when de-watered, becomes an aggregate-like solid ⁹⁸.

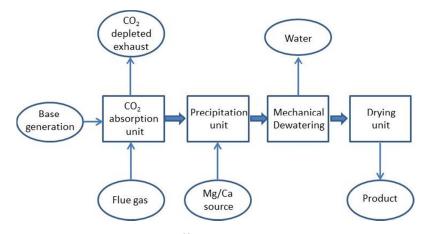


Figure 6. Calera process diagram (modified from ⁹⁸).

Another method of sequestering carbon dioxide using brines, referred as SkyMine, has been recently proposed ¹⁰¹. CO_2 is absorbed into an aqueous caustic soda mixture to form carbonate and/or bicarbonate products ¹⁰¹. Flue gas from power plant is cooled from 300°C to 30°C in a series of heat exchangers and then is introduced at the bottom of an absorber, where NaOH is used. The latter is produced by brine electrolysis. The reaction taking place in the carbonation chamber is:

$$2NaOH + CO_2 = Na_2CO_3 + H_2O \text{ and } Na_2CO_3 + H_2O + CO_2 = 2NaHCO_3$$
 (18)

The carbonate and bicarbonate formed are separated from the liquid solution by heating, which can be done by exchange of heat with the flue gas (93% purity of Na_2CO_3 is achieved) or heat derived from hydrogen produced in electrolysis. In the membrane cell processing units, the following inputs and products are obtained:

At anode: 26% NaCl + 2,275 kWh/t Cl ₂ = Cl ₂ (g) + 24% NaOH	(19)
At cathode: $H_2O + e = 30-33\%$ NaOH + H_2 (g)	(20)

This process claims to reach a conversion of 98% by using large amounts of NaOH and /or electricity ¹⁰¹. In addition to capturing and mineralizing CO₂, the SkyMine process also claims the possibility to clean SOx and NO₂ from the flue gas, and remove heavy metals, such as mercury. A joint venture namely Skyonic Corporation, which includes BP and ConocoPhillips, has started the costruction of a commercial CO₂ capture plant to remove 83,000 t CO₂ per year from a cement plant (130,000 considering the reduced emissions in producing backing soda). The strength of the process is represented by the possibility to produce valuable carbon-negative products (e.g. hydrochloric acid and sodium bicarbonate) using low-cost chemical inputs in a low energy requirement capture-mineralisation plant ¹⁰². Despite the fact that the reliability of the process has been proved and large investment has been received (\$128 millions), a comprehensive cost assessment is not publicaly available. Also, it has to be noticed that market for HCl and sodium bicarbonate are not large for a wide application of this technology.

3.1.2.2.5 Organic acid direct processes

Organic acids and their anions may affect mineral weathering rates by three possible mechanisms: (1) changing the dissolution rate far from equilibrium either decreasing solution pH or forming complexes with cations at the mineral surface , which provides a new parallel reaction mechanism for the detachment of material from the mineral surface; (2) ability to make aqueous complexes with aqueous metals that would otherwise inhibit rates; and (3) changing the ions speciation in solution, which affects the dissolution rate of minerals ^{103,104}. Far from equilibrium, the dissolution rates of most silicate minerals increases exponentially with increasing hydrogen ion concentration (low pH) in solution. The pH effect can be explained by the fact that sorption of protons on an oxide surface polarizes the metal-oxygen bonds, weakening the bonding with the underlying lattice ¹⁰³.

Recently, Bonfils et al. ⁶² have proposed a direct mineral carbonation process where organic acids are used to enhance the dissolution of silicate rocks. The interactions between organic ligands and magnesium silicates have been reported in the geochemical literature since organic acids are more efficient that water in accelerating silicates leaching dissociation of Mg-O-Si bonds in presence of protons ^{59,105}. Bonfils work showed that the presence of disodium oxalate under 20 bar of CO_2 pressure lead to the formation of strong oxalate–magnesium complexes in solution and precipitation of MgC₂O₄·2H₂O (glushinskite), which impede the precipitation of magnesium carbonate. Contrary to oxalate, citrate and EDTA salts ligands do not form any solid by-products with magnesium, but also do not promote carbonation, arising strong doubts on the possibility of developing a direct aqueous mineral carbonation process using organic salts ⁶². Moreover, Declercq et al. ¹⁰⁴ investigated the effect of organic ligands on olivine (forsterite) dissolution at 25°C and pH 3. The study included the evaluation of acetate, oxalate, citrate, EDTA, glutamate, gluconate, malonate, aspartate, tartrate, malate, alginate, salycilate and humate. Their study, in agreement with previous reports, concluded that aqueous organic ligands have at most a small effect on forsterite dissolution rates in strongly acidic conditions but may have an effect at higher pH (4-7).

The contrasting effects of organic acids on steady-state forsterite dissolution rates with increasing pH was related to their aqueous speciation, since these organic species are in form of neutral species at acidic pH, but as negatively charged aqueous species in mild acidic and neutral conditions ¹⁰⁴.

3.1.3. Indirect Carbonation

Indirect mineral carbonation refers to processes that take place in more than one stage. Indirect carbonation typically involves the extraction of reactive components (Mg^{2+} , Ca^{2+}) from minerals, using acids or other solvents, followed by the reaction of the extracted components with CO_2 in either gaseous or aqueous phase. Pure carbonates can be produced using indirect methods, due to removal of impurities in previous carbonate precipitation stages ^{23,106}.

Munz et al. ¹⁰⁷ demonstrated the principles of separating magnesite and silica after dissolution of olivine in carbonated aqueous solutions using a flow-through column reactor. The process consisted of three steps: 1) dissolution of 75 μ m fine olivine at 130°C and 150 bar; 2) precipitation of magnesite at 250°C; and 3) precipitation of silica. Both precipitation steps were dependent on pH and temperature ¹⁰⁷. A carbonation efficiency of 11% and 93% were obtained after mechanical pre-treatment (211 kwh/t) after 2 and 18 hours, respectively ⁶⁰. However, as high carbonation efficiency was only obtained after long times, the process is not viable in an industrial scale.

A wide number of strong acids and bases such as HCl, H_2SO_4 and HNO_3 have been employed for the dissolution of silicate rocks ^{2,23,56,108}. Lin et al ¹⁰⁸ proposed a 2-stage process, where serpentine is decomposed to magnesium hydroxide using HCl at 150°C. The resulting Mg(OH)₂ was then carbonated at 325°C for 2 hours in a fixed bed at atmospheric pressure ¹⁰⁸. However, because the authors did not address the recovery of the chemicals used in the process and the time required for

dissolution was too long compared to the precipitation stage, the process was not attractive. Maroto-Valer et al. ¹⁰⁹ developed a process, where serpentine was chemically activated with H₂SO₄ at temperature 20-65°C for 3-12 hours. The resulting magnesium sulphate was reacted with sodium hydroxide to precipitate $Mg(OH)_2$ following an exothermic reaction. The $Mg(OH)_2$ subsequently reacted with CO_2 in aqueous suspension at 20°C and 40 bars. Conversion of 55% was achieved in 10 minutes under these mild conditions. Sulphuric acid was regenerated by reacting CO₂ with MgSO₄¹⁰⁹. However, chemicals make up and intensive chemical regeneration hindered the deployment of this process. The effect of HCl, H₂SO₄ and HNO₃ on serpentine dissolution at 20°C and different solution concentrations (1, 2, and 4 M) revealed that their capacity in dissolving the mineral decreases in this order: $H_2SO_4 > HCl > HNO_3$. Despite their effectiveness in extracting Mg from silicates, processes that employed strong acids did not result in viable MC processes due to the overall difficult and large energy penalties associated with their recovery ^{2,56}. Organic acids have also been investigated in mineral carbonation to reduce the energy penalty associated to strong acids. Teir et al. ⁵⁶ found that acetic acid (CH₃COOH) and formic acid (HCOOH) were able to leach a significant amount of magnesium from serpentine. Krevor and Lackner ¹¹⁰ established that the sodium salts of citrate, oxalate, and ethylenediaminetetraacetic acid (EDTA) significantly enhance the dissolution of serpentine under weakly acidic conditions. In their process, finely ground serpentine of particle size less than 75µm was reacted in a solution with dissolved salts under a CO2 atmosphere and at 120°C ^{110,111}. This energy penalty can be avoided dissolving the serpentine at an essentially neutral pH, i.e. in a solution more weakly acidic than carbonic acid. The reactions rates were several orders of magnitude higher in the presence of citrate than in the weakly acidic solution alone. Carbonation was performed at 20 bar and 120°C. After 2 hours a conversion of 60% was achieved, while 80% was reached after 7 hours. EDTA, which forms magnesium complexes several orders of magnitude more stable than oxalate and citrate, cannot be used as a catalyst ^{110,111}. The recovery and regeneration of the additive salts were not addressed in this work, which represents a major limit for its potential evaluation, as in the case of strog acids. Also, the long reaction times at the highest CO_2 conversion would require a very large plant footprint to be economic. Succinic acid was also employed to extract reactive component (Ca^{2+}) from wollastonite at 80°C and 30 bar. A promising calcium dissolution of 90% was achieved after 1 hour, but the carbonation step was not tested ¹¹². Park et al. ^{113,114} demonstrated that a mixture of 1 vol% orthophosphoric acid, 0.9 wt.% oxalic acid and 0.1% EDTA greatly enhanced the leaching of magnesium from serpentine at 70°C and 1 bar. After 1 hour dissolution, the slurry was filtered to remove SiO₂ residue. The use of internal agitation with grinding media in the dissolution stage greatly improved the extraction of magnesium from serpentine. The filtrate rich in Mg^{2+} and Fe^{2+} was then carbonated bubbling through CO_2 at ambient temperature. Overall, the conversion achieved was 65%, but recovery of additives was not addressed even in this case ^{113,114}.

With the aim to improve the efficiency of mineral dissolution and recover and re-use additives, Maroto-Valer and co-workers proposed a pH-swing CO₂ mineralisation process using ammonium salts ⁶⁸. At 100°C, 1.4M aqueous solution NH₄HSO₄ was found to extract 100% Mg from serpentine in 3 hours. The proposed process consists of five steps and the main reactions are presented in Figures 7. In the first step, NH₃ was used to capture CO₂ from flue gas to produce NH₄HCO₃. In the mineral dissolution step, 1.4 M NH₄HSO₄ was used to extract Mg from serpentine ground to a particle size range 75-125µm. The Mg-rich solution was then neutralised by adding NH₄OH, after which impurities in the leaching solution were removed by adding NH₄OH. The Mg-rich solution is then reacted with the product from the capture step NH₄HCO₃ to precipitate carbonates. Since formation and stability of hydro-carbonates is temperature dependent, MgCO₃·3H₂O (nesquehonite) can be converted to 4MgCO₃·Mg(OH)2·4H₂O (hydromagnesite) at temperatures above 70°C. Precipitation of hydromagnesite resulted in a solution mainly containing (NH₄)₂SO₄. The final step was the additive regeneration, with the decomposition of (NH₄)₂SO₄ at ~330°C, and producing NH₃ for the capture step and NH₄HSO₄ for the dissolution step ^{68,72}.

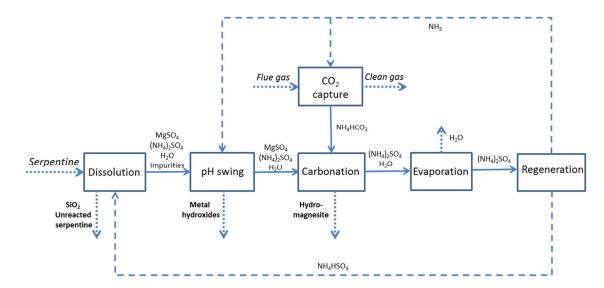


Figure 7. pH-swing CO₂ mineral carbonation process with recyclable ammonium salts (modified from ⁶⁸).

In a typical capture process, the CO₂ is first absorbed by chemicals (e.g. NH₃) and then desorbed (to recover the sorbent) and compressed for transportation, where stripping and compression consumes about 70% of the total CCS energy consumption. Since CO₂ captured as sodium carbonate/bicarbonateis directly used in the proposed mineral carbonation, there is no need for desorption and compression of CO₂. This process as other pH swing processes is also able to separate three different products: silica, magnesite and iron oxide ^{56,66,68,106,114}. This process could also be integrated with the chilled ammonia CO₂ capture process, which has been demonstrated to capture more than 90% of CO₂ (from 3-15% CO₂ in flue gas) ¹¹⁵ and an estimated energy penalty of 477 kWh/t CO₂ ¹¹⁶.

The main drawback of the aqueous pH swing ammonium-based process is represented by the large amount of water that needs to be separated from the salts during the regeneration step. Based on their work, where a solid to liquid (S/L) ratio or 50 g/L was used, 50-56 t H₂O were required to sequester 1 t CO₂. Since water evaporation is a high energy penalty process, they attempted to reduce the water usage in the system ¹¹⁷. When the S/L ratio increased to 300g/L, 16 t H₂O were required to sequester 1 t CO₂. However, since the CO₂ fixation efficiency decreased to 46.6%, larger amount of reactants (serpentine and salts) were required. Moreover, the amount of water to be evaporated is still too high and alternative separation methods need to be investigated in order to make this process economically feasible.

A two-step process which also uses ammonium salts has been recently developed by Zevenhoven and co-workers ^{118,119}. The scheme of this process is shown in Figure 8. In the first step, Mg(OH)₂ was produced from serpentinite and in the second step, Mg(OH)₂ was carbonated in dry phase. This process takes advantage of: (1) the higher reactivity of Mg(OH)₂ compared to that of serpentinite and MgO and (2) the potential recovery of the heat of reaction released during the carbonation. In the Mg extraction step, a mixture of serpentinite was heated together with ammonium sulphate at a temperature of 450-550°C. This resulted in the formation of a magnesium sulphate that was than dissolved in water. Adjusting the pH using ammonium hydroxide or ammonia led to precipitation of Mg(OH)₂, while iron oxide was recovered as by-product. Finally, Mg(OH)₂ was carbonated in a fluidised bed at 20 bar and 500-550°C. A 50% Mg conversion was achieved in 10 minutes ¹²⁰. With this work, the authors attempted to reduce the high energy requirements for the regeneration steps of aqueous pH swing processes (water/salts separation by thermal evaporation) and also attempted the recovering of the carbonation heat. Although it is thermodynamically feasible to recover heat of reaction in the exothermic carbonation stage for the endothermic magnesium extraction, this has not yet been practically demonstrated. Moreover, modest recovery of magnesium (50-60%) is currently limiting this technology ¹²¹. Mg extraction from serpentine is carried out using ammonium sulphate acidic derivatives (ammonium bisulfate and sulfamic acid are formed at temperature > 300°C). However, sulfamic acid volatilizes and/or decomposes at a significant rate by 400°C, which will affect the recovery of Mg ^{121,122}.

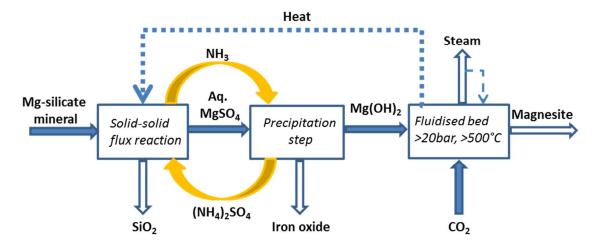


Figure 8. Åbo Akademi mineral carbonation process (modied from ¹²¹).

Hunwick ¹²³ developed a multistep method for capture and sequestration of CO₂ from power station flue gases. This process is being developed by an Australian company, Integrated Carbon Sequestration Pty Ltd, after the proof-of-concept experiments carried out at Commonwealth Scientific and Industrial Research Organisation (CSIRO). In the first step, finely grinded (<40µm) Mg/Ca silicates were mixed with ammonia to produce an aqueous slurry of 30% solids. Then, CO_2 was absorbed into the slurry, which was pumped to a reactor at an elevated pressure of 100 bar and at a temperature of 225° C so as to enhance the rate of a reaction between CO₂ and serpentine. Magnesium carbonate was produced and the ammonia was finally regenerated ¹²³. A pipeline reactor was proposed to transport the slurry produced at the mine to an underground chamber reactor, while another pipe reactor connected the power plant to the mine, where the reaction product would be stored. The proposed carbonation reactor was an underground chamber excavated from bed rock at a depth of 100m sufficient to carbonation reactions to occur ⁹⁶. Despite the fact that this approach looks promising, no data are available on its assessment, including feasibility and energy consumption for the proposed long distance transport to pump the slurries form/to the different locations need to be addressed. Moreover, serious problems could arise from pipeline corrosion as previously reported by O'Connor and co-workers ¹²⁴.

3.2. Technologies developed for waste materials

Some of the drawbacks of mineral carbonation of primary earth minerals could be avoided by using solid wastes generated from large scale industrial processes such as coal or oil shale fired power plant, solid waste incinerator, cement plant, steel and paper industry (Table 5) as a feedstock ^{125,126}. This approach has a number of advantages: (1) these materials are often associated with CO₂ point source emissions; (2) they tend to be chemically less stable than geologically derived minerals ¹²⁷ and thus require a lower degree of pre-treatment and less energy-intensive operating conditions to enhance carbonation yields ^{73,127}; (3) waste materials could supply a readily available source of calcium or magnesium mineral matter (preferably in the form of CaO or Ca(OH)₂) without the need for mining; they are typically fine-grained with high reactive surface areas (CKD, CBD, AODS);(4) hazardous waste can reclassified through pH-neutralization and mineral transformation (MSWI, APC,

asbestos tailings, RM, OS FA), and finally; (5) the end product of the sequestration step may be amendable for re-use in products such as road base or other construction material ¹²⁸ as well as pure and precipitated Ca or Mg-carbonates ^{126,129}. On the other hand, the amounts available of industrial waste materials are relatively limited and rather unpredictable due to developments in technology (changes in availability and chemical composition) and legislation issues ²⁵. Currently, the research has focused on assessing and maximizing the storage of CO₂ by optimizing the operating conditions including pressure, temperature, liquid-to-solid ratio, gas humidity, gas flow rate, liquid flow rate, particle size, and solid pretreatment ^{24,128,130,131}.

The theoretical maximum CO_2 uptake (TCO₂ uptake, Eq.18) of waste expressed in wt.% was calculated using a modified Steinour formula ^{25,132}; ECO₂ indicated the experimental CO₂ uptake.

 $TCO_2 uptake = 0.785 \times (\%CaO - 0.53 \times \%CaCO_3 - 0.7 \times \%SO_3) + 1.091 \times \%MgO + 0.71 \times \%Na_2O + 0.468 \times (\%K_2O - 0.632 \times \%KCl)$ (18)

Table 5 presents a summary of inorganic waste materials (Furnace Slag (BFS), Electric Arc Furnace slag (EAFS), Basic Oxygen Furnace slag (BOFS) Cement Kiln Dust (CKD), Cement Bypass Dust (CBD), Recycled concrete aggregate (RCA), Municipal Solid Waste Incineration ash (MSWI), Air pollution control (APC) residue, Coal and Lignite Fly Ashes (FA), Wood ash (WA), Red Mud (RM), Mine Tailings and Alkaline Paper Mill Wastes Ash (APMWA)), which have been tested as mineral carbonation feedstocks; their % of CaO and MgO, the theoretical and experimental CO₂ capture capacity and the different process conditions investigated.

Table 5. Main properties and carbonation conversions of waste materials (DC – direct carbonation, DAC –
direct aqueous carbonation, IAC – indirect aqueous carbonation, NW – natural weathering, CRW
 – cold rolling wastewater, DI – deionized water)

	CaO, %	MgO, %	TCO ₂ , %	Process	ECO ₂ , %	Remarks
Iron and st	eelmaking sl	ags				
BFS ^{24,25,133}	15-42	5-11	20-44 ²⁵	IAC: 1. step (T=70°C; CH ₃ COOH); 21 step (T= 30°C; P=1bar; NaOH) ¹³⁴	22.7	 +high CO₂ sequestration capacity; + generated in large quantities; +generated near CO₂ source;
				DAC: P=5bar; L/S=0.15; 100% 7 CO ₂ ; t=2h ¹³⁷		+carbonation improves mechanical and environmental parameters of slag;
BOFS 25,133,138	34-56	2-6	29-52 ²⁵	DAC:T=100°C; P=19bar; L(DI)/S=2; 100%CO ₂ ; t=30min; d<38 μm ¹³⁹	25	+possible applications: PCC; -must undergo milling, except AODS and LFS
				DAC: T=70°C, P=1bar; L(DI)/S=10; 100%CO ₂ ; t=2h ¹³¹	22.9	- high T&P, US (€4000/t slag ¹³⁵), or additives are needed for acceptable
				DAC: T=60°C; P=1.47bar; L(DI)/S=20; 100%CO ₂ ; t=30 min	28.9	 conversion; Cost: €77/t-CO₂ net avoided (DAC) ⁷⁰; IAC with PCC production: 300 kWh/t-CO
				DAC:T=25°C; P=1ba; L(CRW)/S=20; t=125min; d<44 μm^{141}	28.3	⁶⁶ or €1990/t-CaCO ₃ for chemicals only ¹³⁶ .
				IAC: 1. step (T= 80° C, NH ₄ Cl; ground d<2000 µm); 2. step (13% CO ₂) ⁶⁶	16.2	-
				IAC: 1. step (T=30°C; CH ₃ COOH); 2. step (T=30°C; P=1 bar; NaOH; 100%CO ₂) ¹²⁶	9	-
EAFS 25,133,138,142	25-47	4-19	24-48 ²⁵	DAC: T=20°C; P=1bar; L(DI)/S=10; 15%CO ₂ ; t=40h; d=38-106µm ¹⁴²	1.74	-

				DAC: T=20°C; P=1bar; L(DI/S=10; 15%CO ₂ ; t=65min; d<100 µm ¹⁴³	1.9-8.7	_
				DAC: P=5bar; L/S=0,15; 100% CO ₂ ; t=2h ¹³⁷	12	_
				DAC: T=50°C; P=3bar; L/S=0,4 ¹⁴⁴	18	
LFS ^{142,143}	42-58	6-15	42 ¹⁴²	DAC: $T=20^{\circ}C$; P=1bar; L(DI)/S=10; 15%CO ₂ ; t=40h; d=38-106 μ m ¹⁴²	24.7	_
				DAC: T=20°C; P=1bar; L(DI/S=10; 15%CO ₂ ; t=65min; ¹⁴³	4.6	_
AODS 133,145	41-61	4-7.5	31-54 25,135	DAC: T=50°C; US; L(DI)/S=10; 100%CO ₂ ; t=4h; d=63-200 μm ¹³⁵	15.1	_
				DAC: T=50°C; US; L(16.6 g/L MgCl)/S=10; 10%CO ₂ ; t=240min 145	27	
				DAC: T=90°C; P=9bar; L/S=16; t=120min ¹⁴⁶	26.4	-
Cement wa	stes					+generated in large quantities;
CKD 25,132,148,149	34-48	1-1.5	10-30	DAC: Ambient T&P L/S=0,33; 5- 15%CO ₂ , 8h ¹³²	8-18	+generated near CO ₂ source (CKD, CBD); + carbonated product can be reused in
				DAC: Ambient T, P=2bar; H_2O to form paste; 100% CO ₂ ; t=72 h ¹⁴⁹	10	cement manufacturing, aggregates, to produce PCC etc;
CBD ^{148,149}	66	1	50	DAC: Ambient T, P=2bar; H_2O to form paste; 100% CO_2 ; t=72 h ¹⁴⁹	25	 +CKD, CBD has fine particle size; -waste cement needs to be ground;
Waste 25-63 cement, RCA ^{150, 151}	25-63	-63 0.3-2	0.3-2 20	$\begin{array}{llllllllllllllllllllllllllllllllllll$	1.6- 16.5	 -low carbon sequestration capacit (CKD); Cost: IAC with PCC production: \$1 323/t-CaCO₃¹⁴⁷
				DAC: $T=20^{\circ}C;$ L(DI)/cement=0.26; 20% CO ₂ ; t=60 min ¹⁵¹	8.9	525/1-CdCO3
				IAC: 1. Step (T=50°C; P=30bar; 100%CO ₂ L/cement=350; d=10- 200 μ m; t=10min). 2 step (T=30°C; 1 bar; t=30min) ¹⁴⁷		_
MSWI ashe	s					+produced in large quantities (MSWI BA);
MSWI BA 25,130,149	22-53	2.8	25	DAC: P=3bar; L/S=0.3-034; 100%CO ₂ , RH=65%; t=2.5h; d<710μm ¹⁵²	3.2	+produced near CO ₂ source; +carbonation reduces pH and leaching of hazardous elements for safer landfill;
				DAC: Ambient T, P=2bar; H_2O to form paste; 100% CO_2 ; t=72 h ¹⁴⁹	4	 +Grinding not required; -low carbon sequestration capacity (MSWI)
APC residue 25,152, 153	36-60	1-2.5	50-58	DAC: P=3bar; L/S=0.2-0.3; 100%CO ₂ , RH=65%; t=2.5h; d<212µm ¹⁵²	7.3	BA)
				$\begin{array}{llllllllllllllllllllllllllllllllllll$	8-12	-
				DC: T= 650-500°C; P=1 bar, 10- 50% CO_2^{67}	25	_
				T= 30-50°C; P=1-10bar, 100% CO ₂ , L/S=0-0.6 ⁶⁷	25	
Fuel combu	stion ashes					
Coal FA 1.3-10 25,154,155	1.3-10	1-3	6-9	DAC: T=20-60°C; P=10-40bar; 100%CO ₂ ; L/S=10; d _{median} =40µm; 18h ¹⁵⁴	2.6	+produced in large quantities (coal FA); +produced near CO ₂ source; _ +grinding usually not required;
				DAC: $T=30^{\circ}C$; $90^{\circ}C$; $P=10-40bar$; L/S=1-10; $100\%CO_2$; ground d=20-150 μ m or <150 μ m ¹⁵⁵	3.6-7.2	 +High CO₂ sequestration capacity (OS FA); -low CO₂ sequestration capacity (Coal FA);
				d=20-150μm or <150 μm ⁻¹⁵⁵		

				DAC: T=90°C; P=40bar; L/S=1; bulk ash; t=2h ¹⁵⁷	6.5	- Waste available in few areas (OS FA); Cost: \$11-21/t-CO ₂ at mineralization	
				NW: ambient T&P wet deposited ash; t=20y ¹⁵⁷	6.8	capacity of 0.1-0.2 t CO ₂ /t-FA 156 .	
Lignite FA	27.5	6.5	43	DAC: T= 75 °C, P=1bar; 10%CO ₂ ; L/S=20; 4,5h; d<250µm ¹⁵⁸	23		
				DAC: T=30-80°C; L/S=40-80; NaCl (1-25g/L; pH=5-9; 100%CO ₂ ; ground d=30-125 μm; t=10- 50min ¹⁵⁹	7,1		
OS FA (PF, CFB) ^{160,161}	38-50	5-12	26-49	DAC: ambient T&P L/S=10; 15%CO ₂ ; t=65min ¹⁴³	29	_	
				NW: ambient T&P wet deposited ash; t=8w ¹⁶¹	2.2		
WA ¹⁴⁹	24-46	8-9	50	DAC: Ambient T, P=2bar; H ₂ O to form paste; 100% CO ₂ ; t=72 h 149	8	_	
Mine tailing	s					+carbonation destroys the asbestos nature	
Asbestos tailings 23,164	0.2	39	43	DC: T=375°C; P=1bar; 56%CO ₂ ; 10% H ₂ O; d=37-75 μm ; t=5h 164	0,5	 (asbestos tailings, Ni tailings if chrysotile present) +grinding not required (asbestos and Ni tailings), +large quantities produced in localized areas; +carbonation stabilizes RM disposal; -low carbon sequestration ECO₂; too expensive to achieve high carbonate conversion (Asbestos and Ni tailings) -bicarbonates generated instead of carbonates (RM) Cost: \$147/t-CO₂ for DAC of RM ¹⁶²; IAC of Ni tailings with hydromagnesite production US\$600-1600/t-CO₂ for chemicals only ¹⁶³. 	
Ni tailings 23,165	3.4	21-40	43	IAC: 1 step (T=70°C; L(4M HCl, HNO ₃)/S=10; t=2h; ground d<0,5 mm); 2. Step (T=30°C; 100%CO ₂ ; NaOH; t=0,5h) ¹⁶⁵	29		
RM ^{23,162,166}	2-7	<1	7-19	DAC: T=20°; P=3.5 bar; L/S=0.2- 0.6; ground d _{mean} =30µm; t=12h	5.3		
				DAC: ambient T&P 100%CO ₂ ; ground d=0,1-160 μ m; 3 carbonation cycles (each 5h) ¹⁶²	7.2		
				DAC: ambient T&P L/S=10; 15%CO ₂ ; d<1000 μm; t=24h ¹⁶⁶	4.15		
Alkaline pape	er mill wastes						
APMWA 148,149,168	45-82	1-5	42-55	DAC:T=30°C; P=10bar; L(DI)/S=20; d=15µm; t=2h ¹⁶⁸	21,8	 +high carbon sequestration capacity; +grinding not required; -generated in small quantities. 	
				DAC: Ambient T, P=2bar; H_2O to form paste; 100% CO_2 ; t=72h ¹⁴⁹	10-26		

3.2.1 Process chemistry and reaction kinetics

Industrial wastes such as APC wastes and ashes from solid fuel combustion often contain a considerable amount of free lime. For aqueous carbonation processes, irreversible hydration of calcium oxide (Eq. 19) is followed by simultaneous dissolution of Ca(OH)₂ (Eq. 20) and dissociation of aqueous CO₂ (Eq. 21, 22) precedes the carbonation reaction (Eq 22) 24,154,169 . As the Ca²⁺-ions are converted to CaCO₃ and precipitated out, more Ca(OH)₂ dissolves to equalize the Ca²⁺ concentration 24 .

$CaO(s) + H_2O(I) \rightarrow Ca(OH)_2(s)$	(19)
$Ca(OH)_2(s) \rightarrow Ca^{2+}(aq) + 2OH^-(aq)$	(20)
$\text{CO}_2(\text{g}) + \text{H}_2\text{O}(\text{I}) \leftrightarrow \text{H}_2\text{CO}_3(\text{aq}) \leftrightarrow \text{H}^+(\text{aq}) + \text{HCO}_3^-(\text{aq})$	(21)
HCO_3^- (aq) + OH^- (aq) $\leftrightarrow CO_3^{2^-}$ (aq) + $H_2O(I)$	(22)
$Ca^{2+}(aq) + CO_3^{2-}(aq) \rightarrow CaCO_3(nuclei) \rightarrow CaCO_3(s)$	(23)

The aqueous carbonation of wastes, in which CaO is bound as a silicate (such as steel slags etc) can in general be expressed by Eqs (21), (24) and (25) 21,139,170 . Firstly CO₂ dissolves in the aqueous phase resulting H⁺-ions (Eq 21). Secondly, Ca (Mg) leaches from the mineral matrix due to slightly acidic environment (Eq 24). Finally, Ca (Mg) carbonate precipitates (Eq 25). The rate and extent of calcium leaching were found to be inversely related to particle size and pH, and increased with increasing temperature, pressure and surface area 24,82,130,171 .

Ca(or Mg)siliscate(s) + 2H ⁺ (aq) \rightarrow Ca ²⁺ (or Mg ²⁺)(aq) + SiO ₂ (s) + H ₂ O(l)	(24)
Ca^{2+} (or Mg ²⁺)(aq) + HCO ₃ ⁻ (aq) \rightarrow Ca (or Mg)CO ₃ (s) + H ⁺ (aq)	(25)

It has also been demonstrated that contaminated solids of cementitious nature can be rapidly remediated whilst binding CO_2 in the process ¹⁷². Carbonation as a stabilization/solidification technique is a process, which capsules toxic waste matter into solid bulk. The reaction products can cause rapid hardening. During carbonation of cementitious materials a sequence of individual steps occurs: (1) CO_2 diffusion in air and (2) permeation through solid is followed by (3) solvation of $CO_2(g)$ to $CO_2(aq)$, (4) hydration of $CO_2(aq)$ to H_2CO_3 , (5) ionization of H_2CO_3 to H^+ , HCO^{3-} and $CO_3^{2^-}$, (6) dissolution of cementitious phases (Ca_3SiO_5 , Ca_2SiO_4) releasing Ca^{2^+} and $SiO_4^{2^-}$ ions, (7) nucleation of CaCO₃ and calcium-silicate-hydrate gel, (8) precipitation of solid phases and (9) secondary carbonation by converting calcium-silicate-hydrate gel ultimately to silicate hydrate gel and CaCO₃ (eq. 26) ^{24,172}. The extent and rate of carbonation depend mainly on diffusivity and reactivity of CO_2 , which in turn depend on binder type and hydration degree as well as pores type and process conditions (CO_2 partial pressure, relative humidity, temperature and pressure) ¹⁷².

 $3CaO \cdot SiO_2 + yH_2O + (3-x)CO_2 \rightarrow (3-x)CaCO_3 + xCaO \cdot SiO_2 \cdot yH_2O + zCO_2 \rightarrow (3-x)CO_2 + yH_2O + yH_2O$

$$(x-z)CaO\cdot SiO_2\cdot yH_2O + zCaCO$$

(26)

Generally, the waste carbonation reaction could occur in four routes: (1) conversion inside solid particle, (2) $CaCO_3$ crystallization on surface, (3) $CaCO_3$ precipitation in bulk solution, and (4) attachment on solid solution ²⁴. According to Huntzinger et. al. ¹³², Huijgen et. al. ^{127,139} and Uibu and Kuusik ¹⁷³, the main mechanisms affecting the rate and extent of carbonation are transportation-controlled mechanisms such as CO_2 and Ca^{2+} -ions diffusion to/from reaction sites, boundary layer effects (diffusion across precipitate coatings on particle surface, dissolution of $Ca(OH)_2$ at the particle surface), and pore blockage/precipitate coating. Typically, the classical shrinking core type model has been used for describing heterogeneous solid - fluid reactions for determination of rate-limiting mechanism ^{132,174-176}.

In the next sections, the carbonation of different inorganic waste materials will be discussed in detail.

3.2.2 Metallurgical slag

Steelmaking processes generate significant amounts of CO_2 (0.28 - 1 t- CO_2 /t-steel ¹⁴²), accounting for 6-7% of global CO_2 emissions ¹³⁸. Also, globally, these processes generate about 315-420 Mt/yr ¹²⁶.

Chemical Society Reviews

Slags form as a result of interactions between process impurities (primarily silica) and lime at various stages of steel production ^{23,133}. The main types of slags produced at steelmaking process are basic oxygen furnace slag (BOF) (62% of total steel slags), electric arc furnace slag (EAF) (29%), and ladle slag (LS) (9%) ^{25,177}. Blast furnace slag (BF) is generated as a by-product from iron production by melting the gangue of the ore, coke ashes and the siliceous and aluminous residues after the reduction and separation of iron from ore ²⁵. Secondary processes for further refinement of stainless steel produce LS and argon oxygen decarburization slag (AOD).

Iron and steel slags consist mainly of Ca-, Mg-, Al-silicates and oxides in numerous combinations ¹³³. Their annual total CO₂ emissions are estimated to be up to 171 Mt of CO₂ ¹³⁴, representing about 0.6% of global CO₂ emissions from fuel combustion 23 . In general steel-making slags require grinding as carbonation pre-treatment 2348 , but the cost of mining and transportation to CO₂ emission sites can usually be avoided. Mineral carbonation of steel slag (Table 5) is in most cases carried out in a water slurry phase (L/S>1 w/w) at ambient ¹⁴¹⁻¹⁴³ or elevated pressure and temperature ^{66,127,139,} ^{140,176,178}. Santos et al. also used ultrasound (US) and/or additives (MgCl) to enhance the carbonation process 135,145 . The CO₂ uptake of slags depends on the operational parameters (temperature, pressure, particle size) similarly to the carbonation of natural Ca-silicates; but it is less energy demanding ^{139,142,176}. As expected, the slags that contain free CaO as opposed to Ca-silicates were more reactive ¹⁴². Calcium from Ca-silicates was leached (Eq 24) after rapid carbonation of free CaO (Eqs 19-23) and then carbonated (Eq. 25) and on the particles' surface. Ca diffusion through the solid matrix was the rate-limiting step due to the formation of a CaCO₃ capsule and Ca-depleted silicate zone 23,139,176 . The ECO₂ values presented a very wide range (1.7 - 28.9%), depending on the type and composition of slag, as well as process conditions. Using elevated pressures and temperatures, additives and US treatment improved significantly the carbonation kinetics ²⁵, but also increased costs (up to €4000/t-slag) ¹³⁵. Particle size was also important variable, as carbonation was significantly improved by using smaller fractions (38-106 µm)^{25,139,142,176}. Estimations have shown that costs of slag (200°C and 20bar of pure CO₂) were €77/t-CO₂ net avoided ⁷⁰. Carbonation of Cacarrying cementitious materials to sequester CO₂ also resulted in the development of high early stage strength for building materials applications, achieving CO_2 uptakes of 7 – 12% in the process ¹³⁷. A possibility to upgrade steel slags into products of high commercial value, such as high-purity precipitated CaCO₃ (PCC), has also been addressed in several studies ^{66,126,133}. A number of extraction agents including HNO₃, H₂SO₄, NaOH ¹³⁸, NH₄Cl ^{66,106}, and CH₃COOH ^{126,133}, CH₃COONH₄, NH₄NO₃ ¹⁰⁶ NH₄HSO₄ ¹⁷⁹ have been investigated for the indirect carbonation route. The use of HNO₃ solution resulted in rapid extraction of Ca and Mg from BOF and EAF slags with CO₂ sequestration capacity of 0.26-0.38 t CO₂/t-slag ¹³⁸. In case of NH₄Cl (at 80°C), 60% of Ca was extracted resulting in PCC of 98% purity. The CO₂ sequestration capacity of 16% (0.16 t CO₂/t-slag) was achieved in the process with a total energy consumption of 300 kWh/t-CO₂ 66 . Acetic acid extraction resulted in 31-86% carbonate conversion and PCC of 99.5-99.8% purity 126 . Weaker acids, elevated temperatures and longer reaction times promoted selective Ca extraction. Production of high-purity PCC using acetic acid route would cost €1990/t-CaCO₃ assuming the byproduct sodium acetate could be sold for ~€680/t 136 . A closed loop multi-step process was developed to extract Ca²⁺ from steel slag with NH₄HSO₄ solution to form solid CaSO₄, which after pH adjustment and precipitation of impurities reacted with $(NH_4)_2CO_3$ (from CO_2 capture with NH_3) to precipitate $CaCO_3$ ^{179, 180}. The carbonation efficiencies achieved by the latter process ranged from 59-74%¹⁸⁰.

3.2.3 Municipal solid waste incinerator ashes

Incineration is a common management option for municipal solid wastes (MSW). Incineration of MSW (MSWI) offers the reduction of mass (up to 70%), volume (up to 90%) and organic content of wastes, as well as disinfection and potential energy recovery 23,130,152,181,182 . It also generates solid residues, ~80% bottom ash (MSWI BA) and ~20% air pollution control (APC) residues, and atmospheric CO₂ emissions 23,25,183 .

3.2.3.1 MSWI BA

MSWI BA is generally classified as a non-hazardous waste 23,130,152 consisting mainly of silicates (SiO₂, Ca₂Al[AlSiO₇]), sulfates (CaSO₄, CaSO₄·2H₂O, Ca₆Al₂(SO₄)₃(OH)₁₂·26H₂O), carbonates (CaCO₃), but also metal oxides, hydroxides (Ca(OH)₂, Fe₂O₃, Fe₃O₄) and chlorides 25,130 . As Ca and Mg contents of MSWI BA are typically too low for significant CO₂ sequestration, the mineral carbonation technique is mainly applied to achieve a chemically stable structure with improved leaching behavior 130,152 for different applications (e.g. secondary building material in road sub-bases, wind and noise barriers, etc) 23,183 .

Mineral carbonation of MSWI BA has been examined and compared to natural weathering to reduce alkalinity and trace metal mobility ^{152,184-186}. Rate-controlling mechanisms and the effect of operating parameters, such as temperature, CO_2 partial pressure, liquid to solid ratio (optimum L/S~0.3 w/w ^{152,182}), residence time, and particle size ^{152,183} have also been investigated. MSWI BA has shown CO_2 uptake on the order of 3.0– 6.5 wt.% ^{130,149,152,183} (Table 5). Baciocchi et al ¹⁸⁷ tested BA from refuse derived fuel (RDF) incineration with CO_2 uptakes ranging from 4-14% depending on particle size.

The mechanisms involved in the carbonation of these materials are complex. Although most of the studies have only considered $Ca(OH)_2$ carbonation ^{71,183}, it is likely that Ca- and Mg-silicates also take part in CO_2 binding reactions. The dissolution of Ca from the solid matrix into the liquid phase and the diffusion of CO_2 into the pores have been reported as the rate-limiting steps ¹³⁰. The kinetics of CO_2 uptake include the two following reaction steps: (1) an initial rapid CO_2 uptake, which involves "faster" reacting minerals such as $Ca(OH)_2$ and followed by (2) a decrease in the rate until an approximately constant value of CO_2 uptake is achieved ^{152,183,184}, which involves less-reactive Ca-Mg-silicates ¹³⁹.

3.2.3.2 APC residues

APC residues are formed in the process of the flue gas treatment and typically contain a mixture of fly ash, unburned carbon and unreacted lime. Due to the lime content (typically pH>12), and high concentration of heavy metals (Zn, Pb, Cd, Cr, Cu, Hg, Ni), soluble salts and chlorinated compounds, APC residues are classified as a hazardous waste 23,25,152 . High percentage of readily-active calcium hydroxides makes the carbonation of APC residues potentially suitable for CO₂ sequestration 152,188 . Also, the APC carbonation products present a pH value that meets the regulatory limits (pH<9.5) 172,189 .

The carbonation route for APC residues is more straightforward as compared to MSWI BA, since the main reactive species include Ca(OH)₂ and CaClOH. The kinetics of CO₂ uptake showed similar trend as for MSWI BA, only with higher weight gains attributed to higher lime contents and larger specific surface areas ¹⁵². The ECO₂ uptake of the APC residues ranges from 7 to 25 wt.% (0.07 – 0.25 t-CO₂/t-APC) ^{25,67,152,153,188} (Table 5). Baciocchi et al. compared dry (1 bar, 10-50% CO₂, 350-500°C) and wet route (1-10 bar, 100% CO₂, 30-50°C, L/S=0-0.6). Although both routes achieved a similar maximum conversion to carbonates (65%), corresponding to at potential CO₂ storage capacity of 0.25 t/t-APC residues, the dry route presented faster reaction kinetics ⁶⁷.

3.2.4. Power plant ashes

Coal fired power plants provide ca 40% of world's electricity ¹⁵⁴, generating annually 12,000 Mt CO₂ and 600 Mt fly ash (FA) ^{23,25}. About 30% of coal FA is utilized for construction materials ²⁵. Coal FA is a fine powder (particle size typically 10–15 μ m), which composition varies depending on the mineral content of fuel. Generally, it consists of amorphous aluminosilicate glass matrix (Si_xAl_yO₂) and recrystallized minerals, including quartz (SiO₂), cristobalite (SiO₂) and mullite (3Al₂O₃2SiO₂) ²⁵. The main components of bituminous coal FA are SiO₂, Al₂O₃, Fe₂O₃, MgO (1–3%) and CaO (5–10%) and various amounts of unburned carbon ²⁵.

FA from oil-shale (OS, low-grade fossil fuel) combustion has been investigated as a potential sorbent for mineral carbonation. OS FA contains 12–30% free CaO, depending on combustion regimes (PF or

CFB) ¹⁶⁹. Also, potential CO_2 sorbents are produced in power plants that co-fire wood and coal. The resulting wood ash (WA) contains about 45% CaO ¹⁴⁹.

The maximum CO₂ sequestration potential of bituminous coal FA is relatively low ~9 wt% ^{25,154}, but it could be as high as 43-49% for Ca-rich lignite type coal or oil shale ashes ^{143,158} (Table 5). Studies have mainly been focused on direct aqueous carbonation route under mild process conditions with either water ^{154,157,158,169} or brine ^{155,190} as the reaction medium or by natural weathering over longer period of time ^{157,161}. A pilot scale mineral carbonation process was developed and tested by reacting coal FA with flue gases in a fluidized bed reactor at a 2120 MW coal-fired power plant in Point of Rocks, USA. According to a preliminary economic analysis of the process reported that 90% CO₂ capture from a 532 MW power plant would cost about \$11-21/t-CO₂ assuming sequestration capacity of 0.1-0.2 t-CO₂/t-FA ¹⁵⁶. According to general estimates, coal FA with an average CO₂ sequestration capacity of 5% could sequester 0.25% of CO₂ emissions from coal fired power plants ^{23,25,154}.

3.2.5. Cement wastes

Cement kiln dust (CKD) is a fine by-product of Portland cement and lime high-temperature rotary kiln production ²⁵. The cement industry generates 0.15–0.20 t of CKD per tonne of cement (world output 2.8 Gt) ^{23,25,132} and 5% of global CO₂ emissions ¹⁹¹. CKD is characterized as a potentially hazardous waste due to its caustic nature ^{23,191} and disposed into appropriate landfills or stockpiles. Typical CKD contains 38-48% CaO and 1.5-2.1% MgO ^{23,25,132} (Table 5), but significant amount of CKD is already carbonated (CKD contains 46-57% CaCO₃ ¹³²). Cement bypass dusts (CBD), which are removed after kiln firing have much lower carbonate content than CKD, and therefore much higher potential to capture CO₂ (0.5 t-CO₂/t-CBD) ¹⁴⁸. The CKD ECO₂ uptake achieved was 8–25 wt.% (i.e. 0.08–0.25 t-CO₂/t-CKD,CBD) at ambient temperature and pressure in a column reactor ¹³² or in a pressurized reactor ¹⁴⁸. According to Bobicki et al. ²³, CKD can potentially sequester up to 42 Mt of CO₂ annually, or about 0.1% of global CO₂ emissions from fuel combustion. Also, as a result of carbonation, potential health hazard associated with CKD disposal are eliminated ¹²⁸.

Waste cement is a by-product from aggregate recycling process, where waste concrete is pulverized and classified to separate the aggregate from the waste cement. According to Bobicki et al ²³, waste cement has a potential to store up to 61 Mt CO_2 considering the annual waste concrete production of 1100 Mt from EU, USA and China together ²³. However, the majority of waste cement is currently already reused in construction applications ²⁵.

Teramura et al. ¹⁵⁰ used a CO₂-activated hardening process to produce building materials, where waste cement was mixed with water (50% H₂O) before moulding it into bricks, curing with CO₂ and drying overnight (with maximum ECO₂ 16.5%, at 100% CO₂ and 4 bars, Table 5). Kashef-Haghigi and Ghoshal ¹⁵¹ achieved carbonation efficiency of 18% and ECO₂ of 8.9% when by curing fresh concrete blocks in a flow-through reactor (20% CO₂ in N₂, 20°C and 60 min). Small demonstration scale are planned for technology of using point source CO₂ emission to limit the need for heat and steam in the curing process in the production of precast concrete products is currently moving towards a small-scale demonstration ¹⁹².

Katsuyama et al. ¹⁴⁷ and lizuka et al. ¹⁷¹ produced high purity $CaCO_3$ using indirect aqueous carbonation route for the extraction of Ca^{2+} from cement waste by pressurized CO_2 (30 bar) and subsequent carbonation at reduced pressures (1 bar). The estimated costs per 1 metric tonne of $CaCO_3$ were US\$136 for desulfurization and US\$323 for high-purity $CaCO_3$ (market price of $CaCO_3$ of about \$400/t CaCO₃¹⁴⁷).

3.2.6. Mining tailings

Asbestos, copper, nickel, platinum deposits, diamondiferous kimberlite pipes and podiform chromite deposits hosted by dunite, serpentinite, and gabbronorite produce tailings suitable for mineral carbonation 23,193 . Also, bauxite residue (red mud), from alumina processing is a suitable feedstock for CO₂ sequestration $^{162, 166, 167}$. In addition to CO₂ storage, the carbonation treatment also improves the

properties of wastes, especially in case of asbestos tailings and red mud, enabling safer landfilling or reuse ²³.

3.2.6.1 Asbestos tailings

Production of 1 t of asbestos (4 Mt globally) generates ca 20 t of tailings ²³. The tailings from chrysotile processing are often associated with residual asbestos and, are therefore classified as hazardous wastes. Carbonation of asbestos tailings could be useful in several ways, as they contain up to 40% MgO, the mining and size reduction are already done and the asbestiform nature of the mineral is destroyed. Thus, both the remediation of a hazardous waste and the sequestration of CO_2 could be achieved ⁸². The natural weathering of old tailings piles have been studied by several groups ^{193,194} (Table 5). Wilson et al. ¹⁹³ estimated that the chrysotile in the tailings piles had carbonated approximately 0.3% per year. It was suggested that the CO_2 sequestration efficiency could be enhanced by optimizing the surface area and particle size of the tailings, as well as bioleaching (adding microorganisms to increase solubilisation of alkaline earth metals) ¹⁹⁵. Larachi et al. ¹⁶⁴ investigated the direct carbonation in dry and humid (humidity 0-10%) environments over a range of temperatures (300-1200°C) and low CO_2 pressures and achieved maximum carbonate conversion of 0.5% after 5 h at 375°C in moist atmosphere. Carbonation of Mg-rich wastes such as asbestos tailings requires elevated pressures and temperatures or pre-treatment, similar to serpentine treatment, in order to achieve higher carbonate conversions.

3.2.6.2 Nickel tailings

As high-grade sulphide deposits are almost depleted and laterites require more complex processing than sulphide ores, the nickel industry has focused on low-grade sulphide resources, often hosted in ultramafic rocks ²³. Processing ultramafic ores generates vast quantities of Mg-rich tailings (40% MgO, Table 5). Valorizing these ultramafic tailings could make marginal nickel projects economically feasible ¹⁹⁶. Teir et al. ^{163,165} extracted magnesium from serpentinite (from stockpile nickel tailings mine) using a variety of acids (HCl, HNO₃) (Table 5). The Mg-extracts were carbonated (carbonate conversion 94%) in a multistage process with a carbonate conversion of 94% and producing individual precipitates of silica, iron oxide and hydromagnesite of 93-99% purity ¹⁶⁵. However, the estimated costs for chemicals only were US\$600-1600/t CO₂ ¹⁶³. Integrating nickel mining operations with CO₂ sequestration requires further developments to reduce costs and a carbon regulatory framework including a cap-and-trade scheme with sufficiently high carbon price ²³.

3.2.6.3 Red mud

Red mud (RM) is the caustic waste material of bauxite ore processing for alumina extraction. Producing 1 t of alumina generates 1.0–1.5 t of highly alkaline RM (70 Mt annually ¹⁹⁷)¹⁶⁷. RM typically contains Fe₂O₃ (30-60%), Al₂O₃ (10-20%), SiO₂ (3-50%), Na₂O (2-10%), CaO (2-8%) and TiO₂ (trace - 10%) ¹⁶², which are present in portlandite (Ca(OH)₂), sodium carbonate (Na₂CO₃), NaAl(OH)₄, Na₆[AlSiO₄]₆, crystalline hematite (Fe₂O₃), goethite (R-FeOOH), gibbsite (Al(OH)₃), boehmite (γ -AlOOH), sodalite (Na₄Al₃Si₃O₁₂Cl), anastase (TiO₂), rutile (TiO₂), katoite (Ca₃Al₂SiO₄(OH)₁₂), gypsum (CaSO₄·2H₂O), and quartz (SiO₂) ¹⁶⁶. Mineral carbonation of RM reduces its toxicity and leaching behavior in terms of long-term storage in addition to CO₂ sequestration ^{167, 197}. Carbonated RM can also be used for various applications such as fertilizers, brick and tile industry, plastics industry, wastewater treatment and cement production ¹⁶⁶. RM is generally carbonated via direct process route at ambient temperatures and pressures ^{162, 166, 167} and sequestration capacity of 0.04 – 0.05 t CO₂/t RM have been reported (Table 5). The roughly calculated cost of CO₂ sequestration is at US\$147/t-CO₂ ¹⁶². At Kwinana in Western Australia, Alcoa operates a residue carbonation plant, where gaseous CO₂ from a nearby ammonia plant is contacted with RM, reducing pH of the slurry to a less hazardous level and capturing in the process 0.030-0.035 t CO₂ per t of RM ¹⁹².

As the primary source of alkalinity in RM is NaOH, the main carbonation products are Na_2CO_3 and $NaHCO_3$ ¹⁶². Soluble Na-carbonates provide a less permanent CO_2 storage than solid Ca-Mg-

carbonates because of their solubility. In order to provide a more permanent CO_2 storage option, RM was mixed with brine solution (solution of hydrated Ca- and Mg- chlorides) prior to carbonation ^{197,} ¹⁹⁸. It has been estimated that over 100 Mt of CO_2 have been sequestered in RM through the natural weathering of historically produced RM (6 Mt annually) ¹⁹⁹. By utilizing appropriate technologies for incorporating binding cations into RM, approximately 6 Mt of additional CO_2 could be sequestered whilst RM is also remediated ¹⁹⁹.

3.2.7. Alkaline paper mill wastes

Cellulose pulp production for paper manufacture results in the formation of several types of alkaline paper mill wastes (APMW), which typically contain 45-82% ^{148,168} free CaO and are therefore suitable sorbents for mineral carbonation (Table 5). The pulp mills also generate CO₂, which could be used to carbonate the APMW. Produced CaCO₃ could be utilized in the pulp and paper industry or sold as a value-added by-product ^{23,168}. DAC of APMW presents ECO₂ of 16-27% (i. e. 0.16-0.27 tCO₂/t-APMW) at 20-30°C and 2-10 bar over 2-72 h ^{149,168}.

4. Products Utilisation

The effective development of utilization routes for the materials produced by mineral carbonation could help to make this technology economically viable and facilitate its deployment. Based on the information discussed in the previous sections, processes that produce multiple separated products are preferred in terms of products utilisation, due to high purity required by the market. Therefore, processes where cations are extracted from the feed material in a separate step (indirect processes) may be suitable for controlling the morphology and particle size of precipitated products for high-value applications, compared to direct mineralisation technologies ^{200,201}. It has been shown that indirect processes are generally able to separate silica, magnesium/calcium carbonate and iron hydroxides with purities as high as 90% by switching the pH of the solution from acidic to basic ^{10,72,200,202}. The dissolution of silicates selectively removes Mg/Ca and other elements from the mineral matrix, leaving behind silica in amorphous phase with particle in the range of tens of microns and purity of about 80% ^{163,200}. Precipitated hydromagnesite (Mg₅(CO₃)₄(OH)₂ 4H₂O) with purity > 93wt.% can be produced by indirect processes 72,163,201 . Also, 99 wt% pure calcite with particles > 5µm have been produced by carbonation of wollastonite under mild conditions (10 bar, 100°C) ²⁰⁰. However, purity required to access high-value markets is difficult to be achieved without extra purification steps ¹⁰.

Applications for carbonate products can be divided in low-end high-volume and high-end low-volume uses. For the MC products to be commercially used there are specifications and quality criteria that must be met (e.g. particle size, distribution and low level of contaminants). Construction and fillers applications seem the most appropriate for silica and carbonates products, respectively, while feedstock for iron/steel works may represent the natural pathway for iron oxides from MC. Among the low end applications, MC products as liming agent to buffer the acidity of soils is promising, but require the MC products to be free from potential pollutants that might derive from particular flue gas or mineral wastes converted in carbonates. Also, land reclamation from the sea in coastal areas and mine reclamation using silica, magnesium and calcium carbonates are other possible low-tech high-volume applications ¹⁰.

High-end applications usually require stringent specifications. Mono-disperse nano-particles uniform in size, shape and composition have a wide number of applications in industry, such as catalysts, chromatography, ceramics, pigments, pharmacy, photographic emulsions, etc ²⁰³.

MC can be used to produce silica in amorphous phase and with particles smaller than <30 μ m, which could serve as a pozzolanic cement replacement material or as a filler ²⁰⁴. Silica from MC, for deoxydiser in steel making, circuit boards, ceramix matrix composites, semiconductors should reach very high purity (SiO₂ > 98.5%; Fe₂O₃ < 0.1%, Al₂O₃ < 0.15%) ^{10,205}. Similar purity is expected for ceramics applications, while slightly lower purity would be required for use a refractory material (95% SiO₂) and iron and steel making (90% SiO₂). It is unlikely that MC products can reach purity

levels required for silicon applications without additional post-processing. Amorphous silica, which is a fine powder and it is currently considered a high quality reactive cement additive, may represent the most likely application for amorphous silica from MC¹⁰.

Calcium carbonate is extensively used as a novel functional material in several fields such as plastics, rubber, paint, printing ink, weaving, toothpaste, make-up, and foodstuffs. Calcium carbonate is a product in MC processes that use inorganic wastes or calcium silicates, such as wollastonite.

An interesting perspective to filler technology is the development of nano-sized, high performance, and low cost fillers from calcium carbonates in forms of ground calcium carbonate (GCC) and precipitated calcium carbonate (PCC)¹⁰. Calcium carbonate can precipitate in six different forms, namely amorphous calcium carbonate (ACC), hexahydrate calcium carbonate (HCC), monohydrate calcium carbonate (MCC) and the polymorphs calcite, aragonite and vaterite, which have trigonal, orthorhombic, and hexagonal crystal system, respectively²⁰⁶. For the PCC applications, several physical and chemical properties, such as particle size average and distribution, morphology, specific surface area, polymorph or the chemical purity are very important in determining the potential market ²⁰⁷. The different polymorphs of CaCO₃ can have different functions as additives. For example, dispersion can be increased if cubic CaCO₃ is added as an addition in paint; acicular or rod-like CaCO₃ has a reinforcing effect on rubber and plastics; and spherical CaCO₃ has a significant impact on the brightness and transparency of ink²⁰⁸.

By controlling the initial concentration of the reagents, stirring speed, pH, type and amount of additives, and other reaction conditions, $CaCO_3$ with different polymorphs, morphologies, and grain sizes can be obtained. For example, different $CaCO_3$ polymorphs were generated by changing the carbonation time or after aging ²⁰⁸.

Particle sizes and morphologies of precipitated CaCO₃ varied from rhombohedral (15–35 nm) to scalenohedral (400 nm in diameter and 2 μ m length) with changing the operating variables, CaO concentration, CO₂ flow rate and in surfactants concentration ²⁰³. Addition of etyltrimethyl ammonium bromide cationic surfactant (2%) produced narrow size rounded particle morphologies either rhombohedral or spherical and limited amount of agglomerate ²⁰³.

Finally, enzimes such as Carbonic Anydrase (CA) has been used to enhance carbonation efficiency and modify the properties of the reaction products. Mesoporous alumina synthesized from egg shell membrane and pore-expanded SBA-15 were used as template to immobilise CA ^{209,210}. The carbonation capacity of alumina immobilized CA was found to be 25% lower compared to that obtained in presence of free CA ²⁰⁹. Favre et al. reported that at higher pH, calcite and vaterite were observed while at lower pH, only calcite was favoured ²¹⁰. Another biomimetic complex (Co-BBP) that mimics the catalytic activity of carbonic anhydrase (CA) in mineral carbonation, was prepared by the coordination of cobalt (II) with 2,6-bis(2-benzimidazolyl) and was encapsulated into a metal organic framework (Co-BBP@Tb-MOF). The biomimetic catalyst enhanced CO₂ hydration and calcium carbonate (CaCO₃) crystallization as CA. The metal organic framework was determined the CaCO₃ morphology, resulting in the formation of circular plate structures ²¹¹.

Producing precipitated calcium carbonate (PCC) from wastes can contribute to the reduction of Wastes that contain high calcium content, such as steel slag can be utilized as PCC if calcium is selectively extracted prior to carbonation to fulfill the requirements of purity and crystal shape. Zevenhoven and co-workers selectively extracted calcium from the slag with aqueous solution of ammonium salt (NH₄NO₃, CH₃COONH₄ or NH₄Cl) producing PCC from the steel slag derived calcium rich solution with properties comparable to the PCC produced with conventional methods. However, a very small liquid ratio (5 g/l) was required to get high dissolution efficiency (73%), rending this method expensive because of the large reactor volumes required ²¹². Also, calcium carbonate powder produced from steel slag presented inferior brightness to traditional PCC (mainly due to iron and manganese content) and resulting in a decrease of the market value of the alternative product. Despite this, the separation of iron oxide before the carbonation stage can enhance the quality of PCC produced by this method ^{200,201,213}. Recently, an innovative synthesis of goethite–calcite nanocomposite has been proposed. This synthesis involved the sequential precipitation of (1)

nanosized acicular goethite (a-FeOOH); (2) the instantaneous precipitation of portlandite (Ca(OH)₂) by adding CaCl₂ salt to a goethite alkaline suspension (2NaOH + CaCl₂ = Ca(OH)₂ + 2NaCl) and; (3) submicrometric calcite precipitation by injection of CO₂ into a goethite–portlandite alkaline suspension (Ca(OH)₂ + CO₂ = CaCO₃ + H₂O)²¹⁴. The precipitated nanocomposite had surface area of around 92 m²/g when synthesized at 30°C and 45 m²/g, when synthesized at 70°C.

It has to be mentioned that other methods have been proposed to convert CO_2 into chemicals and fuels. Compared to utilisation of MC products as construction or filling materials, which could in theory absorb Gt of CO_2 , industrial utilizations of CO_2 as solvent and reactant amount to only 0.5 wt% (128 Mt /y) of the total anthropogenic CO_2 emissions every year, so that it may not necessarily help mitigate the greenhouse effect considering energy input and carbon circulation ²¹⁵. Catalysed hydrogenation or photocatalytic and electrocatalytic conversion of CO_2 to hydrocarbons has been extensively reviewed ²¹⁵⁻²¹⁷. Even if technologies have been developed for large-scale CO_2 hydrogenation to methanol or methane, their deployment is mainly limited by the high price of renewable hydrogen. Instead, significant technical and catalytic advances are still required for the large-scale use of electro- and photocatalytic routes, due to their current low energy efficiency and productivity ²¹⁶.

An indicative order of magnitude of the current and potential future CO_2 consumption is presented in Table 6. EOR and urea yield boosting are technologies already in use. MC technologies, algae cultivation and potentially ECBM could utilize flue gas directly and therefore would not require a conventional capture plant to deliver a concentrated CO_2 stream ¹⁹². A semi-quantitative ranking process identified mineralisation technologies (mineral carbonation and concrete curing), EOR and algae cultivation having the greatest potential to accelerate alternative forms of CCS. This assessment considered a series of 14 criterions including the CO_2 emissions in the act of reusing it ¹⁹².

	Existing (future) CO ₂
CO ₂ Uses	demand, Mt/h
Enhanced oil recovery	30-300 (<300)
Urea	5-30 (<30)
Food and beverage	~17 (35)
Water treatment	1-5 (<5)
Other	1-2 (<6)
Enhanced coal bed	
methane recovery	(30-300)
CO ₂ concrete curing (MC)	(30-300)
Algae cultivation	(>300)
Mineralisation (MC)	(>300)
Red mud stabilisation (MC)	(5-30)
Baking soda (MC)	<1
Liquid Fuels (methanol,	
formic acid)	(>600)

Table 6. Current and potential future CO₂ consumption (modified from ¹⁹²).

5. Summary and remarks

It is generally accepted that to reduce the level of CO_2 emitted in the atmosphere, a portfolio of different and complementary technologies such as renewables, change in energy uses and CCS has to be employed. Mineral carbonation has the potential to sequester billions of tonnes of CO_2 , but the

current costs are too high for a widespread deployment of this technology. This work reviews the current state of mineral carbonation routes and the role they can play in decreasing the emissions of CO_2 .

5.1 MC options comparison

In-situ MC has great potential in terms of volume of CO₂ which could be permanently fixed within the hosting rocks as solid carbonates thus reducing the risk of potential seepage from the storage site. There is a large availability of minerals which can react in-situ with the injected CO₂, both onshore and offshore and often close to anthropogenic sources of CO₂. In-situ MC can also be beneficial for the worldwide development of storage projects. Abundant onshore and offshore basalts and peridotites are available for in-situ low temperature mineralisation. The largest layered onshore basalt formations are located in India (provinces of Deccan Traps), USA (Columbia River basalts), Russia (Siberian Traps) and UAE/Oman²¹⁸. The current limits of in-situ carbonation are due to the slow pace of the process and the need for artificial ways of enhancement of the chemical reactions which require large amount of energy. Identifying specific sites where natural characteristics such as geothermal gradients are favourable to the carbonation process may reduce the associated costs.

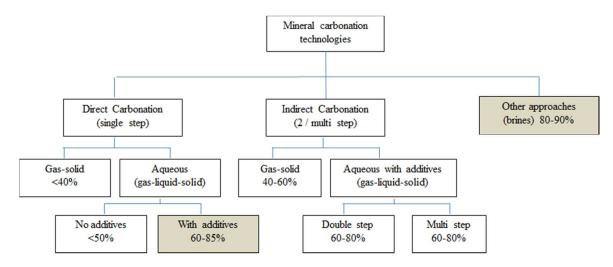
Ex situ MC presents intrinsic materials handling issues, due to the large mineral requirements and associated reaction products, which result in large process scale (larger than actual power plant materials handling), and it seems to be only employable to existing small-medium emitters. MC may be suitable to large emitters if the new plants are designed with the required infrastructures. Since for small-medium emitters, geologic sequestration may not be an economically viable option, and there are no commercialized processes that specifically address this technology gap, MC may target this market. Large ultramafic rock deposits within a 100-200 km radius of power/industrial plants emitting over 1Mt/y CO₂ are available in South Africa, China, Russia, Kazakhstan ²¹⁹, New South Wales in Australia ⁹³, USA and Europe ⁸². However, not all these resources are easily accessible. Mg-bearing silicates such as serpentine and olivine represent the most suitable mineral resources, while other Mg-silicates are less attractive due to their low Mg content and/or low availability.

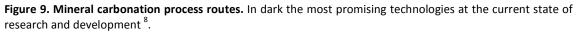
Moreover, a number of large scale industrial wastes can be considered as feedstock for CO_2 mineralisation. Regardless of several benefits, such as avoiding costs for mining and transport, the current CO_2 mineralization technologies developed for wastes still cannot compete with geological storage in terms of potential quantity and cost of sequestrated CO_2 . The ones that appear to carbonate easily in mild conditions (contain free lime, do not require additional grinding, bind CO_2 effectively even from dilute flue gases etc.) and have a high carbon sequestration capacity (APC residues, APMW, OS FA) are only available locally or in too small quantities to make a global impact. However, especially in countries that lack geological storage, these options should be considered (for instance OS FA could capture 10-12% of CO_2 emitted from OS based heat and power sector ¹⁶¹).

Overall, the processes that are attracting major attention (Figure 9) and that seem to be viable at this point have in common the potential production of sellable products, the co-removal of different pollutants from the flue gas and process integration essential to lower the costs. The conceptual integration of high temperature and pressure industrial mineral carbonation facility into a developing mine site has been recently demonstrated to be feasible at an operating cost of ~ $$83/t CO_2^{220}$.

Direct gas-solid processes, which require temperatures up to 500°C and fine grinding of minerals (5-35 μ m) achieve low capture efficiency and are not viable in industrial scale at the current scale of development. On the contrary, it is well documented in the literature that the presence of water considerably enhances the reaction rate in the carbonation process ⁹. Feedstock pre-treatment by fine grinding, thermal activation and chemicals in direct aqueous carbonation processes show significant improvements on CO₂ capture efficiency (up to 85% with pure CO₂ stream) ⁸². Meanwhile, the regeneration and recyclability of additives (NaOH, NaHCO₃) still need to be addressed. The NETL modified processes proposed by Brent and Shell make use of the low grade heat from power plants and from the serpentine thermal-activation to decrease the overall energy consumption. However, not public data are available to estimate the potential deployment and costs associated to these processes.

Multistep aqueous indirect processes in presence of additives are also able to reach high carbonation efficiency using mild process conditions and short residence time as a result of faster reaction kinetic in presence of additives. However, the energy intensive chemicals regeneration step is slowing the development of this group of technologies. Also, the use of catalytic enzymes such as carbonic anhydrase is unlikely to be effective due to its instability and very high costs. At the current state of the art, indirect routes seem to be still too expensive to be competitive as CCS technology for large deployment.





5.2 Demonstration projects

So far, mineral carbonation has been implemented only in few demonstration plants: the first is the Calera process, in the gas fired Moss Landing plant (USA), which has been running for about two years. The plant showed the technical capacity to capture CO_2 (30kt/year) from a 10MW power generator at 90% efficiency, with an associated energy penalty of 10-40%. However, the potential impact on water balances and hydrology from extraction and reinjection of brines and the conclusion that the tested brines (technically unsuitable), sea water (too costly) and alkaline wastes (limited availability) render this process unsuitable for operations at significant scale ⁹⁸.

Another brine based process (Skyonic) is approaching the commercialization stage. Skyonic is currently retrofitting Capitol's cement mill (San Antonio, USA) owned by Capitol Aggregates. This process directly processes flue gas and produces hydrochloric acid, bleach, chlorine, and hydrogen. In terms of large deployment, an evident drawback of this process is going to be the scale of products generated compared to current markets.

Similarly, only few projects based on inorganic wastes have moved to commercial or even small-scale demonstration phase. For example, a pilot scale mineral carbonation process that uses coal FA has been installed at a 2120 MW coal-fired power plant to reduce CO_2 , SO_2 and Hg emissions in Point of Rocks, USA ^{156,221}. Also, accelerated carbonation has been applied for the commercial production of aggregates from APC residues ²²² and in a residue carbonation plant for Red Mud stabilization at Kwinana in Western Australia ⁹⁶.

Carbonation of Red Mud has been run by Alcoa since 2007 locking 70 ktCO₂/y generated in a nearby ammonia plant ¹⁹². However, ~30t red mud/tCO₂ are used, which is about ten times the typical rate of serpentine rock usage. Also, this technology requires a concentrated and preferably high pressure

source of CO₂ (85% pure) to be located in reasonable proximity to an alumina refinery ²²³⁻²²⁵. The CarbFix demonstration project (in-situ MC), where 5% CO₂ in water has been injected in porous basalts near the continental margins has recently shown that is feasible to sequester more than 80% of the CO₂ injected in less than 1 year at 20-50°C ²²⁶. This mineral trapping pathway avoids one of the major drawbacks associated to geological storage in sedimentary basins, since CO₂ dissolved in water is not buoyant and also offers a storage potential one order of magnitude higher than the potential CO₂ emissions from burning all fossil fuels resources ²²⁶.

5.3 MC costs assessment

One of the major challenges for CCS including MC projects is the cost. Recently, an estimated transport and storage cost of 1 (17/tCO₂, which is about double of the cost associated to geological storage in sedimentary basins, has been associated to in-situ MC in basaltic rocks ²²⁶. Therefore, total cost of in-situ MC will be in the range 72-129/tCO₂ (considering a CO₂ capture cost of \$55-112/tCO₂), which is by far larger than recent European carbon market CO₂ price of 2 (7/tCO₂. However, it has to be pointed out that geological storage costs do not taken into account potential long term monitoring costs due to the un-reactivity of dry CO₂ in sedimentary rocks. Also, risk the in-situ MC option drastically reduces potential leakages ²²⁶.

Due to lack of commercial applications, mineral carbonation cost estimates reported in literature are roughly based on laboratory or pilot scale experiments. As expected, the main energy and cost penalties are related to plant size, pre-treatment (grinding feedstock and thermal-treatment), operating conditions (mixing, high temperature/pressure), additives (extraction of reactive species) and separation/disposal of the reaction products.

The most reliable ex-situ mineral carbonation cost evaluation still remains that calculated by the NETL extensive work, since feasibility studies from the other promising technologies are not available. On the basis of these calculations, sequestration costs for a direct process with pre-treatments were estimated to be 50, 90 and 210 $\frac{10}{100}$ for olivine, wollastonite and serpentine, respectively ⁸². Recent developments of the NETL process estimated an overall mineral carbonation process cost of A\$ 70/tCO₂ avoided, where the direct use of thermal heat instead of electrical energy, coupled to partial dehydroxylation with heat integration leaded to a 63% decrease in energy requirement for thermal-activation ⁹¹⁻⁹³.

Hujigen et al ⁷⁰ estimated a sequestration costs (based on depreciation of investments and variable and fixed operating costs) of $\leq 102/t$ CO₂ net avoided for wollastonite.

The cost of direct aqueous carbonation of inorganic wastes such as concrete waste and steel slag was estimated to be in the range of US\$8 -104 /t-CO₂, depending on the operation conditions (spray trickle bed systems in air ²²⁷ or at 200°C, 20bar, 100% CO₂ ⁷⁰).

Using indirect aqueous carbonation route with production of value-adding product such as highpurity PCC or hydromagnesite from EAF slags or serpentine would require chemicals (HCl, HNO₃, CH₃COOH, NaOH) for a cost of \$600-4500/t-CO₂ if not regenerated. An acid (HCl) extraction route technology assessment was performed by IEA GHG (2000), concluding that the calculated cost of €179/t CO₂ avoided made this approach unattractive. A similar conclusion was reached by Teir et al.¹⁶³ using HCl/HNO₃ for the dissolution of the feedstock and NaOH in the precipitation step. They found that the regeneration of the chemicals used would emit 2.6–3.5 times the amount of CO₂ bound in the carbonation process ^{136,163}. CO₂ sequestration using IAC of cement waste at 30 bar and 50° C would require \$136-323/t-CaCO₃, depending on product purity ¹⁴⁷.

As already discussed, these costs are still higher compared to geological storage cost. However, costs of CCS by geological storage in industrial plants such as refineries are less well known and could lead to higher costs due to their complexity (e.g. Mongstad in Norway). For example, a capture costs applied to chemicals, fertilizers, refineries and gas fuelled plants could be up to \$235/tCO₂due to geographic location, production/operating specifics and new technology versus retrofit capture situations ²²⁹. Also, MC does not present the same uncertainties of geological storage in terms of potential leakages and potential long term monitoring costs. In order to compete with geological

Chemical Society Reviews

storage, CO₂ sequestration by MC must offer some additional economic benefits including remediation or stabilization of the hazardous wastes such as asbestos tailings, nickel tailings, and red mud, MSWI and power plant ashes or production of value adding products (building materials, feedstock for existing processes). Recently, the production of artificial aggregates from CO₂ carbonation has been demonstrated by a UK based spin-out company of the University of Greenwich, Carbon8. These aggregates have been further used to produce carbon neutral building block (CO₂, sand, cement) manufactured by Ligancite, UK.

Overall, the global low and high-value market for the raw commodities, primarily cement additives, fillers and iron ore feedstock which could be produced by rock and/or industrial waste mineralisation, is about 27.5 Gt and can be easily flooded assuming 10% of the global CO₂ emissions sequestered by MC. However, carbonation technologies, which produce building materials or aggregates, still need to be demonstrated at a scale sufficient to prove their commercial viability in large scale ²³⁰). From the technologies reviewed above, it emerges that the processes that have advanced at the demonstration phase are those that use alkalinity generated by electrolysis of brines, saltwater or alkaline wastes as feedstock. This can be explained by the similarities of these processes to conventional technologies that use brines, by the flexibility in terms of feedstocks (different alkaline sources such as varies brines, sea water, NaOH) and products and by the by the compatibility of the produced materials to existing markets (e.g. aggregates and bricks for Carbon8/Ligancite, feedstock for the Solvay process for Skyonic). Table 7 summarises the primary benefits from the use of CO_2 in terms of CO_2 avoided, energy required to obtain the carbonation products and market values. Although this "advanced" carbonation processes are viewed as an attractive concept, beneficial uses for carbon dioxide are very far from the scale of anthropogenic emissions of this greenhouse gas and therefore suggest to maintaining primary focus on large-scale capture and geologic storage and further develop the carbonation concepts able to produce materials for the larger construction market. However, the CO_2 sequestration potential of wastes remains marginal in global scale of CO₂ emissions.

Carbonation process	Amount of CO ₂ utilised	Value of by- products (\$/tCO ₂)	Energy penalty for by-product process	CO ₂ emissions avoided	Products market	Market size billion \$/y
Skyonic	Cl ₂ : 14 Mt/y; Na ₂ CO ₃ : 20 Mt/y; H ₂ : 836 Mt/y;	Na ₂ CO ₃ : ~300 \$/t, H ₂ : ~10 \$/t, Cl ₂ : 240 \$/t	20%	2.9t/tCO ₂ captured	Solvay process (Na ₂ CO ₃ or CaCO ₃)	3.4-9
Calera	Sand and Aggregate market: 1500 Mt/y; Cement: 24 Mt/y	Aggregate: 7 \$/t, Cement: 100 \$/t	8-28%	0.5t/tCO ₂ captured	CaCO₃ for cement, aggregates	21
Alcoa	2-23 Mt/y	10-300 \$/t	n.a.	n.a.	n.a.	~500

 Table 7. Potential uses of carbonation products from some MC processes ²³⁰.

5.4 Final Remarks

Despite the large resources available for CO_2 sequestration and the clear advantages over geological storage, the costs of both in-situ and ex-situ MC are currently too high for a large deployment of the technology and new systems are being investigated to attempt to overcome the unchanged technology challenges: 1-process energy economics, 2- chemical reaction rates and 3- materials handling issue (for ex-situ carbonation). The current technology research and developments gaps that have to be addressed to enhance the understanding on mineral carbonation and its deployment are as follows:

- Scarce representative raw materials comparison
- Processes performance data incomplete and inaccurate
- MC Integration with point source not well explored

- Incomplete information on cost/energy balance for thermal activation
- Insufficient knowledge of indirect carbonation fundamentals
- Insufficient knowledge of carbonation fundamentals using flue gas
- Lack of assessed reactor technology options and cost studies. A more systematic approach in costing the process should be addressed for comparison purpose
- Process scale and materials handling issue not well explored
- Scarce data on environmental impact of large mining operations

While it may not be a complete solution in itself for large emitters (excluding the favourable cases where for example a large deposit of silicates is closely located to a large emitter), ex-situ mineral carbonation with inorganic wastes could be part of an integrated approach to carbon sequestration, which combines remediation of hazardous wastes such as asbestos tailings and use of readily available fine industrial wastes such as EAF and cement-kiln dusts to meet CO₂ emission goals. On the contrary, in-situ carbonation may be viable for large scale emitters if the current limitations are overcome. However, at these MC technology costs, its deployment as CCS option requires strong financial incentives.

Acknowledgements

The authors thank the Energy Technologies Institute (ETI) that commissioned and funded the work as part of its CCS programme. Also, the authors thank the Centre for Innovation in Carbon Capture and Storage, Heriot-Watt University (EPSRC Grant No. EP/F012098/2) for support and logistics.

References

1 P. Falkowski, R.J. Scholes, E. Boyle, J. Canadell, D. Canfield, J. Elser, N. Gruber, K. Hibbard, P. Högberg, S. Linder, F.T. MacKenzie, B. Moore, T. Pedersen, Y. Rosenthal, S. Seitzinger, V. Smetacek, W. Steffen, *Science*, 2000, **290**, 291-296.

2 IPCC, 2007, R.K. Pachauri, and A. Reisinger, (Eds.), IPCC, Geneva, Switzerland. pp. 104.

3 S. Pacala and R. Socolow, Science, 2004, 305, 968-972.

4 M.E. Boot-Handford, J.C. Abanades, E.J. Anthony, M.J. Blunt, S. Brandani, N. Mac Dowell, J.R. Ferńandez, M.-C. Ferrari, R. Gross, J.P. Hallett, R.S. Haszeldine, P. Heptonstall, A. Lyngfelt, Z. Makuch, E. Mangano, R.T.J. Porter, M. Pourkashanian, G.T. Rochelle, N. Shah, J.G. Yao and P.S. Fennell, *Energy & Environmental Science*, 2013, DOI: 10.1039/c3ee42350f

5 B.Praetorius, K. Schumacher, Energy Policy, 2009, 37, 5081-5093.

6 A.P. Hallenbeck and J.R. Kitchin, Ind. Eng. Chem. Res. 2013, 52, 10788–10794.

7 W. Seifritz, Nature, 1990, 345, 486.

8 J. Sipilä, S. Teir and R. Zevenhoven, Åbo Akademi University report VT 2008-1, 2008.

9 IPCC, 2005. Metz, B., O. Davidson, H. C. de Coninck, M. Loos, and L. A. Meyer (eds.). Cambridge University Press, Cambridge, United Kingdom and New York, NY, USA, 442 pp.

10 A. Sanna, M.R. Hall, M.M. Maroto-Valer, Energy and Environmental Science, 2012, 5, 7781-7796.

11 G. Holmes and D.W. Keith, Phil. Trans. R. Soc. A, 2012, 370, 4380-4403.

12 K.Z. House, A.C. Baclig, M. Ranjan, E.A. van Nierop, J. Wilcox, H.J. Herzog, *Proc. Natl. Acad. Sci.* U.S.A. 2011, **108** (51), 20428–20433.

13 D.S. Goldberg, K.S. Lackner, P. Han, A.L. Slagle and T. Wang, *Environ. Sci. Technol.*, 2013, **47**, 7521-7529.

14 M.D. Eisaman, K. Parajuly, A. Tuganov, C. Eldershaw, N. Chang and K.A. Littau, *Energy & Environmental Science*, 2012, **5**, 7346-7352.

15 G.H. Rau, *Environ. Sci. Technol.*, 2011, **45**, 1088–1092.16 P. Renforth and T. Kruger, *Energy Fuels*, 2013, **27**, 4199–4207.

17 S.S. Salek, Ro. Kleerebezem, H.M. Jonkers, G.-J. Witkamp and M.C.M. van Loosdrecht, *Trends in Biotechnology*, 2013, **31**, 139-146.

18 M. Verduyn, H. Boerrigter, R. Oudwater, G.A.F. van Mossel. 5th Trondheim CCS,16-17th June 2009.

19 M. Verduyn, H. Geerlings, G. van Mossel, S. Vijayakumari. Energy Procedia. 2011, 4, 2885–2892.

20 H. Yan, J.Y. Zhang, Y.C. Zhao, C.G. Zheng, *Science China Technological Sciences*, 2013, **56**, 2219-2227.

21 A. Olajire, Journal of Petroleum Science and Engineering, 2013, 109, 364-392

22 R. Zevenhoven, J Fagerlund, J.K. Songok, *Greenhouse Gases: Science and Technology*, 2011, **1**, 48-57.

23 E. R. Bobicki, Q. Liu, Z. Xu and H. Zeng, *Progress in Energy and Combustion Science*, 2012, **38**, 302-320.

24 S.-Y. Pan, E. E. Chang and P.-C. Chiang, Aerosol and Air Quality Research, 2012, 12, 770–791.

25 A. Sanna, M. Dri, M. R. Hall and M. Maroto-Valer, Applied Energy, 2012, 99, 545-554.

26 A. Kirchofer, A. Becker, A. Brandt and J. Wilcox, Environ. Sci. Technol. 2013, 47, 7548–7554.

27 A-h. A. Park, P. Kelemen, J. Matter, G. Gadikota, Geo-Chemo-Mechanical Studies for Permanent Storage of CO₂ in Geologic Formations, DE-FE0002386, Columbia University, New York, U.S. Department of Energy National Energy Technology Laboratory Carbon Storage R&D Project Review Meeting, August 21-23, 2012. <u>http://www.netl.doe.gov/</u>.

28 R.Berner, A Lasaga., R. Garrels, American Journal of Science, 1983, 283, 641-683

29 S. Gislason, E. Oelkers, E. Eiriksdottir, M. Kardjilov, G. Gisladottir, B. Sigfusso, A. Snorrason, S. Elefsen, J. Hardardottir, P. Torssander, N. Pskarsson, *Earth and Planetary Science Letters* 2009, **277**, 213-222

30 B. M. Reinhardt, Schweizer. Mineral. Petrogr. Mitt., 1969, 49, 1-30

31 F. Boudier and R. G. Coleman, Journal of Geophysical Research, 1981, 86, 2573-2592

32 P. Kelem, and J. Matter, PNAS ,2008, 105/45, 17295-17300

33 A. Paukert, J. Matter, P. Kelemen, E. Shock, J. Havig, Chemical Geology, 2012, 330/331, 86-100

34 G. D. Nicholls, Mineralogical Magazine 1965, 34/268, 373-388

35 C. Dessert, B. Dupre', J. Gaillardet, L. Francois, C. Allegre, Chemical Geology, 2003, 202, 257-273

36 E. Rosenthal, Journal of Hydrology 1987, 89/3-4, 329-352

37 J. Matter and T. Takahashi, Geochemistry Geophysics Geosystems, 2007, 8/2, 1-19.

38 K. House, D. Schrag, C. Harvey, K. Lackner, PNAS, 2006, 103, 12291-12295.

39 P. Olsen, D. Kent, B. Cornet, W. Witte, R. Schlische, Geol. Soc. Of America Bull. 1996, 108/1, 40-77

40 B. McGrail, A. Spane, E. Sullivan, D. Bacon, G. Hund, Energy Procedia 2011, 4, 5653-5660

41 N. Zakharova, D. Goldberg, C. Sullivan, M. Herron, J. Grau, *Geochemistry Geophysics Geosystems*, 2012, **13/11**, 1-22

42 T. Flaathen, S. Gislason, E. Oelkers, A. Sveinbjornsdottir, Applied Geochemistry, 2009, 24, 463-474

43 S. Gislason, D. Wolff-Boenisch, A. Stefansson, E. Olelkers, E. Gunnlaugsson, H. Sigurdardottir, B. Sigfusson, W. Broecker, J. Matter, M. Stute, G. Axelsson, T. Fridriksson, *International Journal of Greenhouse Gas Control*, 2010, **4**, 537-545

44 J. Matter, and P. Kelemen, Nature Geoscience, 2009, 2, 837-841

45 J. Matter, S. Broecker, S. Gislason, E. Gunnlaugsson, E. Oelkers, M. Stute, H. Sigurdardottir, A. Stefansson, H. Alfredsson, E. Aradottir, G. Axelsson, B. Sigfusson, D. Wolff-Boenisch. *Energy Procedia* 2011, **4**, 5579-5585.

46 B. Sigfusson, 8th CO₂GeoNet Open Forum, S. Servolo, Italy 2013.

47 D.Becker, EOS Trans. Am. Geophys. Union, 1998, 79, 369-378.

48 D. Goldberg, T. Takahashi, A. Slagle, PNAS, 2008, 105/29, 9920-9925

49 P. Olsen, J. Smoot, P. Le Tourneau, *Fieldtrip for the 11th Annual Continental Scientific Drilling Workshop, Arlington, 2007, 1-45*

50 D. Folger, J. Hathaway, R. Christopher, P. Valentine, S. Poag, US Geological Survey Circ., 1978, **773**, 1-28

51 T. Maguire, R. Sheridan, R. Volkert, *Geodynamics*, 2004, **37**, 457-485

52 D. Goldberg, D. Kent, P. Olsen, PNAS, 2010, 107/4, 1327-1332

53 D. Goldberg and A. Slagle, Energy Procedia, 1, 2009, 3675-3682

54 O'Connor, W K, Dahlin, D C, Rush, G E, Gerdemann S J, Penner L R, and D. N. Nilsen, Albany Research Centre. DOE/ARC –TR-04-002, USA, 2005.

55 Levenspiel, O., 1972. Chemical Reaction Engineering, second ed. John Wiley & Sons, New York.

56 Teir, S., Revitzer, H., Eloneva, S., Fogelholm, C.-J., Zevenhoven, R., *International Journal of Mineral Processing*. 2007, **83**, 36-46.

57 T. A. Haug, R. A. Kleiv, I. A. Munz, Applied Geochemistry, 2010, 25, 1547–1563.

58 M. Hänchen, S. Krevor, M. Mazzotti, K. Lackner, *Chemical Engineering Science*, 2007, **62**, 6412-6422.

59 M. Hänchen, V. Prigiobbe, G. Storti, T.M. Seward, M. Mazzotti, *Geochimica et Cosmochimica Acta*. 2006, **70**, 4403-4416.

60 T.A. Haug I.A. Munz and R.A. Kleiv, Energy Procedia, 2011, 4, 5029-5036.

61 W.K. O'Connor, D.N. Nielsen, S.J. Gerdemann, G.E. Rush, R.P. Waltera and P.C. Turner, 2001. 18th Annual International Pittsburgh coal conference, Newcastle, Australia.

62 B. Bonfils, C. Julcour-Lebigue, F. Guyot, F. Bodénan, P. Chiquet, F. Bourgeois, *International Journal of Greenhouse Gas Control*, 2012, **9**, 334–346.

63 D. Daval, I. Martinez, J. Corvisier, N. Findling, B. Goffé, F. Guyot, *Chemical Geology*, 2009, 262, 262–277.

64 G. J. Stockmann, D. Wolff-Boenisch, S.R. Gislason, E.H. Oelkers, *Chemical Geology*, 2013, **337–338**, 56–66.

65 R.M. Santos, T. Van Gerven, Greenhouse Gases: Science and Technology, 2011, 1, 287-293.

66 S. Kodama, T. Nishimoto, N. Yamamoto, K. Yogo and K. Yamada, *Energy*, 2008, **33**, 776-784.

67 R. Baciocchi, G. Costa, A. Polettini, R. Pomi and V. Prigiobbe, *Energy Procedia*, 2009, 1, 4851-4858.

68 X. Wang X, M.M. Maroto-Valer, Fuel, 2011, 90, 1229-1237.

69 W.K. O'Connor, D.C. Dahlin, D.N. Nilsen, R.P. Walters, P.C. Turner, 2000, Proceedings of the 25th International Technical Conference on Coal Utilization & Fuel Systems, Clear Water, FL.

70 W. J. J. Huijgen, R. N. J. Comans and G.-J. Witkamp, *Energy Conversion and Management*, 2007, **48**, 1923-1935.

71 S. Arickx, T. Van Gerven and C. Vandecasteele, *Journal of Hazardous Materials*, 2006, **B137**, 235-243.

72 X. Wang, M.M. Maroto-Valer, Chem. Sus. Chem., 2011, 4, 1291-1300.

73 W. J. J. Huijgen and R. N. J. Comans, *ECN-C-03016* ECN-C-03-016, Energy Research Centre of the Netherlands, 2003.

74 K.S. Lackner, D.P. Butt and C.H. Wendt, *Energy Conversion and Management*, 1997, **38**, S259-S264.

75 H.F. DaCosta, M. Fan, A.T. Russell, Patent n. 20100221163, 2010.

76 S. Kwon, M. Fan, H.F.M. DaCosta, A.G. Russell, *Journal of Environmental Sciences*, 2011, 23, 1233-1239.

77 S. Kwon, PhD dissertation, 2011, Georgia Institute of Technology.

78 J.H. Kwak, J.Z. Hu, D.W. Hoyt, J.A. Sears, C. Wang, K.M. Rosso and A.R. Felmy, *J. Phys. Chem.*, 2010, **114**, 4126-4134.

79 J.H. Kwak, J.Z. Hu, R.V.F. Turcu, K.M. Rosso, E.S. Ilton, C. Wang, J.A. Sears, M.H. Engelhard, A.R. Felmy, D.W. Hoyt, *International Journal of Greenhouse Gas Control*, 2011, **5**, 1081-1092

80 W.J.J. Huijgen, G.J. Ruijg, R.N.J. Comans, Geert-Jan Witkamp, Ind. Eng. Chem. Res., 2006, 45, 9184–9194.

81 T.A. Haug, Dissolution and carbonation of mechanically activated olivine: Investigating CO₂ sequestration possibilities, PhD Thesis, Norwegian University of Science and Technology, Trondheim, March 2010, ISBN 978-82-471-1961-7.

http://www.diva-portal.org/smash/get/diva2:303780/FULLTEXT01.pdf

- 82 S. J. Gerdemann, W. K. O'Connor, D. C. Dahlin, L. R. Penner and H. Rush, *Environmental Science and Technology*, 2007, **41**, 2587-2593.
- 83 O.S. Pokrovsky, J. Schott, *Geochim. Cosmochim. Acta*, 1999, **63**, 881-897.

84 W.J.J. Huijgen and R.N.J. Comans, ECN-C--05-022, 2005, Energy Research Centre of The Netherlands, Petten, The Netherlands.

85 W.K. O'Connor, D.C. Dahlin, G.E. Rush, S.J. Gerdemann, L.R. Penner and R.P. Nielsen, 2004, Albany Research Center, DOE/ARC-TR-04-002, Albany, OR, USA.

86 S.J. Gerdemann, D.C. Dahlin, W.K. O'Connor, 2002, 6th International Conference on Green House Gas Control Technologies, Kyoto, Japan.

87 M. Fabian, M Shopska, D. Paneva, G. Kadinov, N. Kostova, E. Turianicova, J. Briancin, I. Mitov, R.A. Kliev, P. Balaz, *Minerals Engineering*, 2010, **23**, 616-620.

88 P. Balaz, E. Turianicova, M. Fabian, R.A. Kliev, J. Briancin, A. Obut, *International Journal of Mineral Processing*, 2008, **88**, 1-6.

89 Chizmeshya et al., 2002, 2nd Annual Conference on Carbon Capture & Sequestration, Pittsburgh.

90 P. Raschman, A. Fedoročková, G. Sučik, Hydrometallurgy, 2013,139, 149-153.

91 H. Boerrigter, Patent no. WO2008142025 A2, 2008.

92 A.Fedoročkova, M. Hreus, P. Raschman, G. Sučik, Miner. Eng., 2012, 32, 1-4.

93 R. D. Balucan, B. Z. Dlugogorski, E. M. Kennedy, I. V. Belovac, G. E. Murch, *International Journal of Greenhouse Gas* Control, 2013, 17, 225-239.

94 A. Sanna, X. Wang, A. Lacinska, M. Styles, T. Paulson, M. M. Maroto-Valer, *Minerals Eng.*, 2013, 49, 135–144.

95 G.F. Brent, Patent No. 20090305378, 2009.

96 GCCS, 2011, Accelerating the Uptake of CCS: Industrial Use of Captured Carbon Dioxide, P. Brinckerhoff, pp. 279.

97 Calera, 2013, Calera process, http://www.calera.com/index.php/technology/the_science/.

98 S.O. Andersen, D. Zaelke; O. Young, H. Ahmadzai, F. Anderson, M. Atkinson, E. Carson, R.J. Carson, S. Christensen, J.S.J. Van Deventer, S. Hanford, V. Hoenig, A. Miller, M. Molina, L. Price, V. Ramanathan, H. Tope, J. Wilkinson, M. Yamabe. Scientific Synthesis of Calera Carbon Sequestration and Carbonaceous By-Product Applications, Consensus Findings of the Scientific Synthesis Team, January 2011,

http://www.igsd.org/climate/documents/Synthesis_of_Calera_Technology_Jan2011.pdf

99 B.R. Constantz et al, Patent No. US8062418 B2, 2011.

100 B.R. Constantz et al, Patent No. US2009001020 A1, 2009.

101 J.D. Jones, Patent No. 7,727,374, 2010.

102 Skyonic, 2013, http://skyonic.com/wp-content/uploads/2010/02/Skyonic-Groundbreaking-Release-September-30-2013.pdf

103 J.I. Drever, L.L. Stillings, *Colloids and Surfaces A: Physicochemical and Engineering Aspects*, 1997, **120**, 167-181.

104 J. Declercq O. Bosc E.H. Oelkers, Applied Geochemistry, 2013, 39, 69-77.

105 A.A. Olsen, J.D. Rimstidt, Geochimica et Cosmochimica Acta, 2008, 72, 1758-1766.

106 S. Eloneva, A. Said, C.-J. Fogelholm and R. Zevenhoven, Applied Energy, 2012, 90, 329-334.

107 I.A. Munz, J. Kihle, Ø. Brandvoll, I. Manchenbach, J.W. Carey, T.A. Haug, H. Johansen and N. Eldrup, *Energy Procedia*, 2009, **1**, 4891-4898.

108 P.C. Lin, C.W. Huang, C.T. Hsiao and H. Teng, Environ. Sci. Technol., 2008, 42, 2748-2752.

109 M.M. Maroto-Valer, D.J. Fauth, M.E. Kuchta, Y. Zhang and J.M. Andrésen, *Fuel Process. Technol.* 2005, **86**, 1627-1645.

110 S.C.M. Krevor, K.S. Lackner, Energy Procedia, 2009, 1, 4867-4871.

111 S.C.M. Krevor, K.S. Lackner, International Journal of Greenhouse Gas Control, 2011, 5, 1073-1080.

112 J. Baldyga, M. Henczka and K. Sokolnicka, Materials letters, 2010, 64, 702-704.

113 A-H. A. Park, R. Jadhav, L-S. Fan, *The Canadian J of Chem. Eng*, 2003, 81, 885-890.

114 A-H. A. Park and L-S. Fan, Chemical Engineering Science, 2004, 59, 5241-5247.

115 E. Gal, Patent No. WO/2006/022885, 2006.

116 A. Kothandaraman A, 2010, PhD Thesis submitvted at Massachusetts Institute of Technology, http://sequestration.mit.edu/pdf/Anusha_Kothandaraman_thesis_June2010.pdf

117 X. Wang, M.M. Maroto-Valer, *Energy*, 2013, **51**, 431-438.

118 J. Fagerlund, E. Nduangu, R. Zevenhoven, *Energy procedia*, 2011, 4, 4993-5000.

119 J. Fagerlund, E. Nduangu, I. Romão, R. Zevenhoven, *Energy*, 2012, **41**, 184-191.

120 J. Fagerlund, J. Highfield and R. Zevenhoven, RSC Advances, 2012, 2, 10380-10393.

121 J. Highfield, H.Q. Lim, J. Fagerlund and R. Zevenhoven, RSC Advances, 2012, 2, 6535-6541.

122 E. Nduagu, T. Björklöf, J. Fagerlund, J. Wärnå, H. Geerlings, R. Zevenhoven, *Mineral Engineering*, 2012, **30**, 75-87.

123 R. Hunwick, Patent No. AU2008/000232, 2008.

124 W.K. O'Connor, D.C. Dahlin, D.C. Nilsen, S.J. Gerdemann, G.E. Rush, L.R. Penner, R.P. Walters, P.C. Turner, 2002. 27th International Technical Conference on Coal Utilization & Fuel Systems, Clearwater, FL, USA.

- 125. J.-H. Wee, *Applied Energy*, 2013, **106**, 143-151.
- 126. S. Eloneva, S. Teir, J. Salminen, C.-J. Fogelholm and R. Zevenhoven, *Ind. Eng. Chem. Res.*, 2008, **47**, 7104–7111.
- 127. W. J. J. Huijgen and R. N. J. Comans, *Environmental Science and Technology*, 2006, **40**, 2790-2796.
- 128. D. N. Huntzinger, J. S. Gierke, L. L. Sutter, S. K. Kawatrad and T. C. Eisele, *Journal of Hazardous Materials*, 2009, **168**, 31-37.
- 129. O. Velts, M. Uibu, J. Kallas and R. Kuusik, Journal of Hazardous Materials, 2011, 195, 139-146.
- 130. G. Costa, R. Baciocchi, A. Polettini, R. Pomi, C. D. Hills and P. J. Carey, *Environ Monitoring* Assess, 2007, **135**, 55-75.
- 131. E. E. Chang, C.-H. Chen, Y.-H. Chen, S.-Y. Pan and P.-C. Chiang, *Journal of Hazardous Materials*, 2011, **186**, 558-564.
- 132. D. N. Huntzinger, J. S. Gierke, K. Kawatra, T. C. Eisele and L. L. Sutter, *Environ. Sci. Technol.*, 2009, **43**, 1986-1992.
- 133. S. Teir, S. Eloneva, C.-J. Fogelholm and R. Zevenhoven, *Energy*, 2007, 32, 528-539.
- 134. S. Eloneva, S. Teir, J. Salminen, C.-J. Fogelholm and R. Zevenhoven, *Energy*, 2008, **33**, 1461-1467.
- 135. R. M. Santos, D. François, G. Mertens, J. Elsen and T. Van Gerven, *Applied Thermal Engineering*, 2013, **57**, 154-163.
- S. Eloneva, Doctoral Dissertation Doctoral Dissertation, Aalto University School of Science and Technology, 2010. https://aaltodoc.aalto.fi/bitstream/handle/123456789/4876/isbn9789526034577.pdf?seque nce=1
- 137. S. Monkman and Y. Shao, *Journal of Materials in Civil Engineering*, 2006, **18**, 768-776.
- 138. F. J. Doucet, *Minerals Engineering*, 2009, **23**, 262-269.
- 139. W. J. J. Huijgen, G.-J. Witkamp and R. N. J. Comans, *Environmental Science and Technology*, 2005, **39**, 9676-9682.
- 140. E. E. Chang, S.-Y. Pan, Y.-H. Chen, C.-S. Tan and P.-C. Chiang, *Journal of Hazardous Materials*, 2012, **227–228**, 97-106.

- 141. E. E. Chang, A.-C. Chiu, S.-Y. Pan, Y.-H. Chen, C.-S. Tan and P.-C. Chiang, *International Journal* of Greenhouse Gas Control, 2013, **12**, 382-389.
- 142. D. Bonenfant, L. Kharoune, S. Sauve, R. Hausler, P. Niquette, M. Mimeault and M. Kharoune, *Ind. Eng. Chem. Res.*, 2008, **47**, 7610-7616.
- 143. M. Uibu, R. Kuusik, L. Andreas and K. Kirsimäe, *Energy Procedia*, 2011, **4**, 925-932.
- 144. R. Baciocchi, G. Costa, E. Di Bartolomeo, A. Polettini and R. Pomi, *Greenhouse Gases: Science and Technology*, 2011, **1**, 312-319.
- 145. R. M. Santos, M. Bodor, P. N. Dragomir, A. G. Vraciu, M. Vlad and T. Van Gerven, *Minerals Engineering*, 2014, 59, 71-81.
- 146. R. M. Santos, J. Van Bouwel, E. Vandevelde, G. Mertens, J. Elsen and T. Van Gerven, International Journal of Greenhouse Gas Control, 2013, **17**, 32-45.
- 147. Y. Katsuyama, A. Yamasaki, A. Iizuka, M. Fujii, K. Kumagai and Y. Yanagisawa, *Environmental Progress*, 2005, **24**, 162-170.
- 148. P. J. Gunning, C. D. Hills and P. J. Carey, *Waste Management*, 2009, **29**, 2722-2729.
- 149. P. J. Gunning, C. D. Hills and P. J. Carey, *Waste Management*, 2010, **30**, 1081-1090.
- 150. S. Teramura, N. Isu and K. Inagaki, *Journal of Materials in Civil Engineering*, 2000, **12**, 288-293.
- 151. S. Kashef-Haghighi and S. Ghoshal, *Industrial & Engineering Chemistry Research*, 2009, **49**, 1143-1149.
- 152. M. Fernández Bertos, X. Li, S. J. R. Simons, C. D. Hills and P. J. Carey, *Green Chemistry*, 2004, 6, 428-436.
- 153. J. Sun, M. Fernandez Bertos and S. J. R. Simons, *Energy and Environmental Science*, 2008, **1**, 370-377.
- 154. G. Montes-Hernandez, R. Perez-Lopez, F. Renard, J. M. Nieto and L. Charlet, *Journal of Hazardous Materials*, 2009, **161**, 1347–1354.
- 155. M. G. Nyambura, G. W. Mugera, P. L. Felicia and N. P. Gathura, *Journal of Environmental Management*, 2011, **92**, 655-644.
- 156. K. J. Reddy, S. John, H. Weber, M. D. Argyle, P. Bhattacharyya, D. T. Taylor, M. Christensen, T. Foulke and P. Fahlsing, *Energy Procedia*, 2011, **4**, 1574-1583.
- 157. G. N. Muriithi, L. F. Petrik, O. Fatoba, W. M. Gitari, F. J. Doucet, J. Nel, S. M. Nyale and P. E. Chuks, *Journal of Environmental Management*, 2013, **127**, 212-220.
- 158. M. Back, M. Kuehn, H. Stanjek and S. Peiffer, *Environmental Science & Technology*, 2008, **42**, 4520-4526.
- 159. M. C. Mayoral, J. M. Andrés and M. P. Gimeno, *Fuel*, 2013, **106**, 448-454.

- 160. R. Kuusik, M. Uibu and K. Kirsimäe, *Oil Shale*, 2005, **22**, 407-420.
- 161. M. Uibu, M. Uus and R. Kuusik, *Journal of Environmental Management*, 2009, **90**, 1253-1260.
- 162. R. C. Sahu, R. K. Patel and B. C. Ray, Journal of Hazardous Materials, 2010, 179, 29-34.
- 163. S. Teir, S. Eloneva, C.-J. Fogelholm and R. Zevenhoven, *Applied Energy*, 2009, **86**, 214-218.
- 164. F. Larachi, I. Daldoul and G. Beaudoin, *Geochimica et Cosmochimica Acta*, 2010, **74**, 3051-3074.
- 165. S. Teir, R. Kuusik, C.-J. Fogelholm and R. Zevenhoven, *International Journal of Mineral Processing*, 2007, **85**, 1-15.
- 166. D. Bonenfant, L. Kharoune, S. b. Sauve, R. Hausler, P. Niquette, M. Mimeault and M. Kharoune, *Ind. Eng. Chem. Res.*, 2008, **47**, 7617-7662.
- 167. V. S. Yadav, M. Prasad, J. Khan, S. S. Amritphale, M. Singh and C. B. Raju, *Journal of Hazardous Materials*, 2010, **176**, 1044-1050.
- 168. R. Perez-Lopez, G. Montes-Hernandez, J. M. Nieto, F. Renard and L. Charlet, *Applied Geochemistry*, 2008, **23**, 2292-2300.
- 169. M. Uibu, O. Velts and R. Kuusik, Journal of Hazardous Materials, 2010, 174, 209-214.
- 170. A. A. Olajire, *Energy*, 2010, **35**, 2610-2628.
- 171. A. lizuka, M. Fujii, A. Yamasaki and Y. Yanagisawa, Ind. Eng. Chem. Res., 2004, 43, 7880-7887.
- 172. B. Fernández Bertos, S. J. R. Simons, C. D. Hills and P. J. Carey, *Journal of Hazardous Materials*, 2004, **B112**, 193-205.
- 173. M. Uibu and R. Kuusik, *Oil Shale*, 2009, **26**, 40-58.
- 174. D. Ki Lee, Chemical Engineering Journal, 2004, 100, 71-77.
- 175. S.-M. Shih, C. u.-S. Ho, Y.-S. Song and J.-P. Lin, *Industrial & Engineering Chemistry Research*, 1999, **38**, 1316-1322.
- 176. S. N. Lekakh, C. H. Rawlins, D. G. C. Robertson, V. L. Richards and K. D. Peaslee, *Metall and Materi Trans B*, 2008, **39**, 125-134.
- 177. C. S. Gahan, M. L. Cunha and Å. Sandström, *Hydrometallurgy*, 2009, **95**, 190-197.
- 178. J. K. Stolaroff, G. V. Lowry and D. W. Keith, *Energy Conversion and Management*, 2005, **46**, 687-699.
- 179. M. Dri, A. Sanna and M. M. Maroto-Valer, *Fuel Processing Technology*, 2013, **113**, 114-122.
- 180. M. Dri, A. Sanna and M. M. Maroto-Valer, *Applied Energy*, 2014, **113**, 515-523.
- 181. L. Wang, Y. Jin and Y. Nie, *Journal of Hazardous Materials*, 2010, **174**, 334-343.

- 182. X. Li, M. Fernandez Bertos, C. D. Hills, P. J. Carey and S. Simon, *Waste Management*, 2007, **27**, 1200-1206.
- 183. E. Rendek, G. Ducom and P. Germain, Journal of Hazardous Materials, 2006, B128, 73-79.
- 184. T. Van Gerven, E. Van Keer, S. Arickx, M. Jaspers, G. Wauters and C. Vandecasteele, *Waste Management*, 2005, **25**, 291-300.
- 185. J. A. Meima, R. D. van der Weijden, T. T. Eighmy and R. N. J. Comans, *Applied Geochemistry*, 2002, **17**, 1503-1513.
- 186. A. Polettini and R. Pomi, *Journal of Hazardous Materials*, 2004, **B113**, 209-215.
- 187. R. Baciocchi, G. Costa, E. Lategano, C. M. Polettini, R. Pomi, P. Postorino and S. Rocca, *Waste Management*, 2010, **30**, 1310-1317.
- 188. R. Baciocchi, A. Polettini, R. Pomi, V. Prigiobbe, V. Von Zedwitz and A. Steinfeld, *Energy and Fuels*, 2006, **20**, 1933-1940.
- 189. H. Ecke, *Waste Management*, 2003, **23**, 631-640.
- 190. Y. Soong, D. L. Fauth, B.H. Howard, B. H., R. J. Jones, D. K. Harrison, A. L. Goodman, M. L. Gray and E. A. Frommell, *Energy Conversion and Management*, 2005, **47**, 1676-1685.
- 191. D. N. Huntzinger and T. D. Eatmon, *Journal of Cleaner Production*, 2009, **17**, 668-675.
- 192. Accelerating the uptake of CCS: industrial use of captured carbon dioxide, Parsons Brinckerhoff and Global CCS Institute, 2011. http://www.globalccsinstitute.com/publications/accelerating-uptake-ccs-industrial-usecaptured-carbon-dioxide
- 193. S. A. Wilson, G. M. Dipple, I. M. Power, J. M. Thom, R. G. Anderson, M. Raudsepp, J. E. Gabites and G. Southam, *Economic Geology*, 2009, **104**, 95-112.
- 194. H. C. Oskierski, B. Z. Dlugogorski and G. Jacobsen, *Chemical Geology*, 2013, **358**, 156-169.
- 195. Y. W. Chiang, R. M. Santos, A. Monballiu, K. Ghyselbrecht, J. A. Martens, M. L. T. Mattos, T. V. Gerven and B. Meesschaert, *Minerals Engineering*, 2013, **48**, 116-125.
- 196. M. Hitch, S. M. Ballantyne and S. R. Hindle, *International Journal of Mining, Reclamation and Environment*, 2009, **24**, 64-79.
- 197. R. Dilmore, P. Lu, D. Allen, Y. Soong, S. Hedges, J. K. Fu, C. L. Dobbs, A. Degalbo and C. Zhu, *Energy & Fuels*, 2007, **22**, 343-353.
- 198. M. Johnston, M. W. Clark, P. McMahon and N. Ward, *Journal of Hazardous Materials*, 2010, **182**, 710-715.
- 199. C. Si, Y. Ma and C. Lin, *Journal of Hazardous Materials*, 2013, **244–245**, 54-59.

- 200 S. Teir, J. Kettle, A. Harlin, J. Sarlin, Proceedings of the 3rd International Conference on Accelerated Carbonation for Environmental and Materials Engineering (ACEME10, Turku, Finland, Nov.29- Dec. 1, 2009, 63-74).
- 201 S. Teir, S. Eloneva, R. Zevenhoven, Energy Conversion and Management, 2005, 46, 2954–2979.
- 202 A-H. Park, L-S.Fan, Patent No. US7, 722,842 B2, 2010.
- 203 S.M. El-Sheikh, S. El-Sherbiny, A. Barhoum, Y. Deng, *Colloids and Surfaces A: Physicochem. Eng. Aspects*, 2013, **422**, 44-49.
- 204 P. Gronchi, T. De Marco, L. Cassar, Patent No. US6,716,408 B1, 2004.
- 205 A. Sanna, Chapter 6, p.199-208, in: Fossil Fuels: Sources, Environmental Concerns and Waste Management Practices, 2013, R. Kumar Ed., Nova Science Pub. Inc., New York, p. 307.
- 206 R. E. F. Lindeboom, I. Ferrer, J. Weijma, J. B. van Lier, *Water Research*, 2010, 47, 3742-3751.
- 207 C.Y. Tai, F.B. Chen, AIChE Journal, 1998, 1790-1798.
- 208 G. Li, Z. Li, H. Ma, International Journal of Mineral Processing, 2013, 123, 25-31.
- 209 S. Wanjari, C. Prabhu, N. Labhsetwar, S. Rayalu, *Journal of Molecular Catalysis B: Enzymatic*, 2013, **93**, 15-22.
- 210 N. Favre, M.L. Christ, A.C. Pierre, J. Mol. Catal. B: Enzym., 2009, 60, 163-170.
- 211 P. C. Sahoo, Y. N. Jang, S. W. Lee, Journal of Crystal Growth, 2013, 373, 96–101.
- 212 A. Said, H-P. Mattila, M. Järvinen, R. Zevenhoven, Applied Energy 2012, 112, 765–771.
- 213 M. Faatz, F. Gröhn, G. Wegner, Adv. Mater., 2004, 16, 996–1000.
- 214 G. Montes-Hernandez , F. Renard, R. Chiriac, N. Findling, J. Ghanbaja, F. Toche, *Chemical Engineering Journal*, 2013, **214**, 139-148.
- 215 W. Wang, S. Wang, X. Ma and J. Gong, Chem. Soc. Rev., 2011, 40, 3703-3727.
- 216 E.V. Kondratenko, G. Mul, J. Baltrusaitis, G.O. Larrazábal and J. Peréz-Ramírez, *Energy Environ. Sci.*, 2013, **6**, 3112-3135.
- 217J. Qiao, Y. Liu, F. Hong and J. Zhang, Chem. Soc. Rev., 2014, 43, 631-675.
- 218 W.S. Broecker, Elements, 2008, 4, 4295-297.
- 219 F. Bodénan, F. Bourgeois, C. Petiot, T. Augé, B. Bonfils, C. Julcour, F. Guyot, A. Boukary, J. Tremosa, A. Lassin, P. Chiquet, 4th International Conference on Accelerated Carbonation for Environmental and Materials Engineering, 10-12 Apr 2013, Leuven, Belgium.
- 220 M. Hitch, G.M. Dipple, *Minerals Engineering*, 2012, **39**, 268-275.
- 221 K. J. Reddy, B. Reynolds and M. D. Argyle, in Proceedings of the 4th international Conference on Accelerated Carbonation for Environmental and Materials Engineering, Leuven, Belgium, 2013, pp. 175-183.
- 222 P. J. Gunning, C. D. Hills and P. J. Carey, in Proceedings of the 4th International Conference on Accelerated Carbonation and Materials Engineering Leuven, Belgium, 2013, pp. 185-192.

- 223 G. Jones, G. Joshi, M. Clark and D. McConchie, Environmental Chemistry, 2006, 3, 297–303.
- 224 B.E.H. Jones and R.J. Haynes, *Critical Reviews in Environmental Science and Technology*, 2011, **41**, 271-315.
- 225 D.J. Cooling, In A.B. Fourie and R.J. Jewell (Eds.), Paste 2007—Proceedings of the Tenth International Seminar on Paste and Thickened Tailings (pp. 3–15). Australian Center for Geomechanics, Perth, Australia, 2007.
- 226 S.R. Gislason and E.H. Oelkers, Science, 2014, 344, 373-374.
- 227 J. K. Stolaroff, G. V. Lowry and D. W. Keith, *Energy Conversion and Management*, 2005, **46**, 687-699.
- 228 GCCSI, Strategic Analysis of the Global Status of Carbon Capture and Storage Report 2: Economic Assessment of Carbon Capture and Storage Technologies. 2009. Global CCS Institute, GPO Box 828, Canberra ACT 2601 Australia.
- 229 IMC, 2008, Alberta CO₂ Capture Cost Survey and Supply Curve, prepared for the Alberta CCS Development Council. Ian Murray and Co. Ltd. <u>http://www.canadiancleanpowercoalition.com/</u>
- 230 CSLF, 2012, Annual Meeting Documents Book, CSLF-T-2012-10, 4 September 2012, Perth, Au.