

A review of mineral carbonation technologies to sequester CO₂

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Carbon dioxide (CO₂) capture and sequestration includes a portfolio of technologies that can potentially sequester billions of tonnes of CO₂ 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 in 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 per tCO₂). *Ex situ* mineral carbonation has been demonstrated on pilot and demonstration scales. However, its application is currently limited by its high costs, which range from \$50 to \$300 per tCO₂ sequestered. Energy use, the reaction rate and material 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 to date indicates 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 adaptation technologies are urgently required.¹ Extensive evidence on the anthropogenic cause of climate change can be obtained from 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 CO₂ at some stage from processes such as combustion for power generation, cement manufacture, iron and steel making and

natural gas treatment. Then, the captured CO₂ 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}

CO₂ capture at power plants and other large point sources represents the most likely tool for the reduction of current CO₂ emissions from fossil fuel use. CCS is not a new concept and a large number of different CO₂ 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 kW h per tCO₂ (2001) to about 200–300 kW h per 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 kW h per 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 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 CO₂ are delaying the deployment of geological storage.⁴ Overall CO₂ geological storage

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poses a great deal of uncertainty in terms of quantification of storage potential, monitoring injected CO₂ and engineering challenges to ensure that the injected CO₂ 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 naturally occurring silicate rocks and has been proposed as

an alternative approach for CO₂ sequestration since the 1990s.⁷ Some MC technologies have recently approached the commercial stage. MC is defined as the reaction of metal oxide bearing materials with CO₂ to form insoluble carbonates:



This reaction can take place either below (*in situ*) or above (*ex situ*) ground. *In situ* mineral carbonation involves the injection of CO₂ into underground reservoirs to promote the reaction between CO₂ and alkaline-minerals present in the geological formation to form carbonates.⁸ *Ex situ* mineral carbonation



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Aimaro Sanna did his PhD in Environmental Engineering at the University of Nottingham working on catalytic processes towards the conversion of biomass to bio-fuels. He then became a Research Fellow at the Department of Chemical and Environmental Engineering, University of Nottingham (2010–2012), working on CO₂ sequestration by mineral carbonation. Since July 2012, he has been working on the development of clean fossil fuel technologies at the School of

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G. Caramanna

Giorgio Caramanna is a geologist with fourteen years of experience including hydrogeology, geomorphology, volcanology, geochemistry, CO₂ geological storage and natural hazards. His most recent activity, built on the experience gained during his PhD studies, focuses on the development of multidisciplinary techniques for the assessment of the impact of CO₂ emissions on the environment aimed to CCS leakage risk management. He has developed specific

laboratory equipment and experiments which are coupled with the study of areas of natural CO₂ emissions as “field tests” to validate the results of the laboratory work. He has published about fifty papers on his research activity.



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systems with the aim of finding out the new utilization areas for Estonian natural resources (oil shale, phosphorites, limestones-dolomites etc.) as well as for inorganic wastes including the ones from the oil shale processing industry – ashes from heat and power production as well as solid residues formed by retorting local fuels. Efforts are focused on the abatement of acidic emissions and on diminishing the negative environmental impact of alkali oil shale processing residues as well.



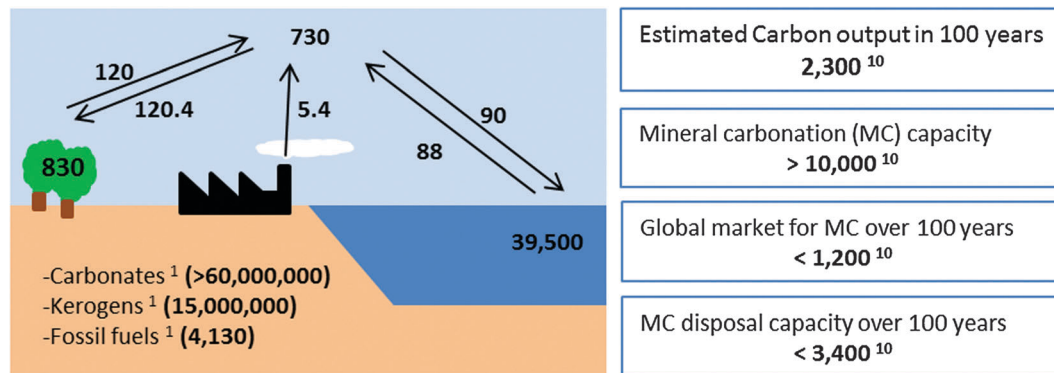
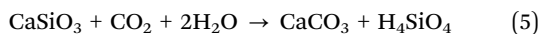
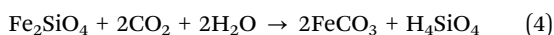
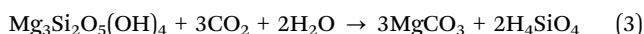
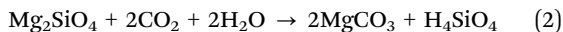


Fig. 1 Global carbon reservoirs (GtC) and net fluxes (GtC per year) (modified from ref. 1, 10 and 17).

relates to above-ground processes, which requires rock mining and material comminution as MC pre-requisites.⁹

MC can take advantage of different starting materials, which include Mg-silicate minerals and Ca²⁺ or Fe²⁺-rich silicates. The reactions occurring in MC processes are listed below.⁹



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 (Fig. 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 CO₂ sequestration potential (>10 000 Gt C) due to the large abundance of silicates around the world, as shown in Fig. 2. MC could also be sourced by CO₂ extracted directly from the air or the ocean. A method to extract CO₂ 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 \$60 per tCO₂.¹¹ However, other works have quantified the costs of this option to be as high as \$600–1000 per tCO₂.¹² Moreover, Goldberg *et al.* estimated that ~75 Mt CO₂ per year could be collected using air capture powered by wind energy and sequestered below seafloor in basalts formation at Kerguelen (see also Section 2.2), where regional reservoirs could hold over 1500 Gt CO₂, sequestering a large fraction of 21st century emissions.¹³ In addition, CO₂ has been extracted from seawater. A total of 59% of dissolved inorganic CO₂ in seawater has been extracted using bipolar membrane electrodialysis with an energy input of ~1527 kW h per tCO₂.¹⁴ While the extraction of CO₂ from air and seawater has been demonstrated, many challenges remain, including slow extraction rates, poor CO₂ selectivity and high costs.

One of the issues associated with 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 Fig. 1, the disposal capacity considering mine and land reclamation projects around the world is considered feasible since large reclamation projects can involve 100–200 Gt 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 an additional environmental impact on marine life, such as localized elevated pH or co-dissolution of trace metals.¹⁶



M. M. Maroto-Valer

Prof. M. Mercedes Maroto-Valer is the Robert Buchan Chair in Sustainable Energy Engineering and Director of the Centre for Innovation in Carbon Capture and Storage (CICCS) at Heriot-Watt University. She is also Head of the Institute for Mechanical, Processing and Energy Engineering. She has established an international research reputation at the interface between energy and the environment and has over 340 publications. Her team is develop-

ing solutions to meet the worldwide strive for cost-effective and environmentally-friendly energy, with emphasis on clean energy technologies, including carbon dioxide capture, transport, storage and utilization.



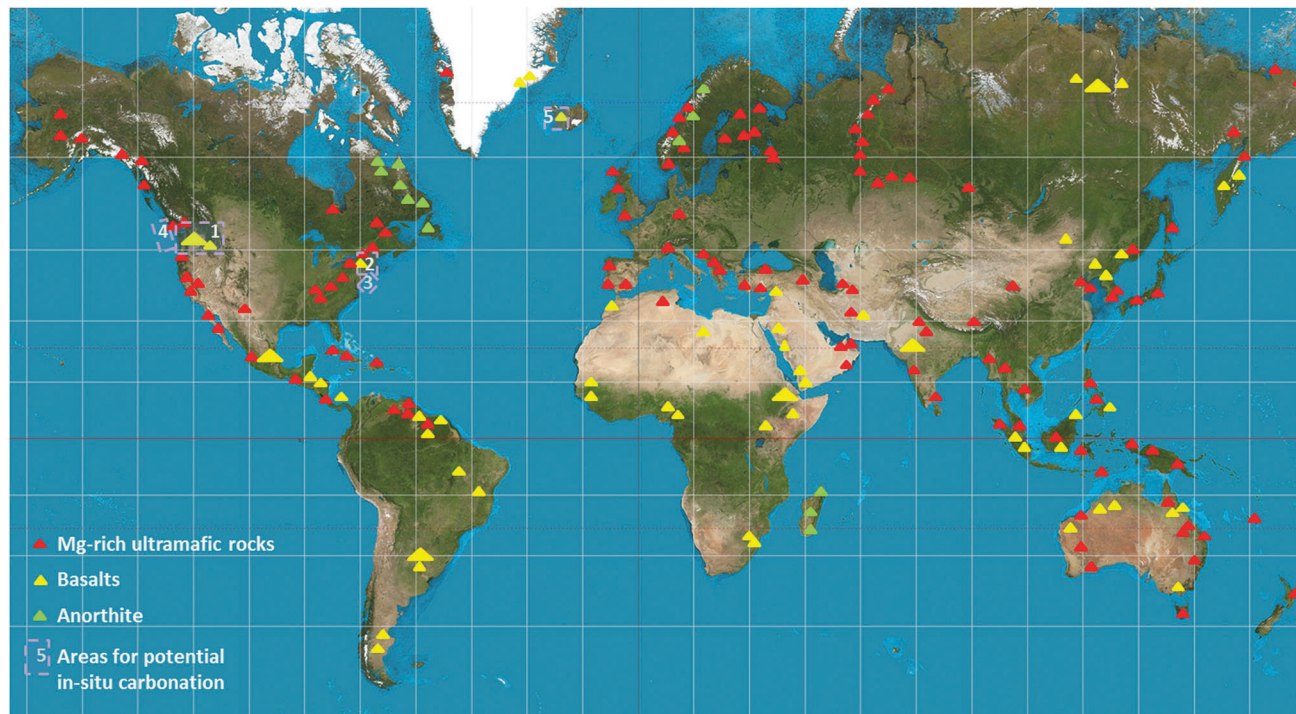


Fig. 2 Mineral feedstock available for MC (modified from ref. 27).

Mineral resource availability, scalability, applicability to regions without geologic storage capacity, inherent stability of the reaction products and the potential revenue from MC products support the on-going development of this technology. Also, mineral carbonation can operate on flue gases directly, without CO₂ pre-separation, which typically stands for 70–75% of the cost of the CCS chain.^{18,19} A CO₂ carbonation efficiency of ~20% has been reported when SO_x and NO_x were present in the flue gas (15% CO₂) using wollastonite at 40 bar and 150 °C.²⁰ However, very few works have been published on MC in the presence of impurities to fully assess this option.

MC effectiveness for CO₂ mitigation purposes has been limited due to the slow kinetics of the CO₂-silicate reactions, energy intensive pre-treatments, logistic issues (*e.g.* locations of mineral resources and CO₂ 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} Olajire²¹ 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-workers¹⁷ 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 CO₂ 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 carbonation of alkaline wastes have been recently published,^{23,24} providing an overview of the types of industrial 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 the presence of silicate rocks and alkaline wastes and product applications. Technologies using mineral rocks as feedstock are treated in more depth, since previous studies clearly indicate that alkaline wastes represent a niche application and would only be able to marginally reduce the global CO₂ emissions (<1%).^{25,26}

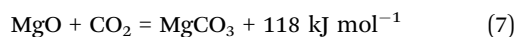
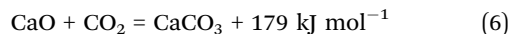
In situ mineral carbonation is analysed in the first section of the work, followed by the evaluation of the silicate rocks for *ex situ* mineral carbonation technologies based on minerals



and inorganic waste feedstocks. Finally, a section on the potential products from mineralisation and final remarks is presented.

2. *In situ* mineral carbonation

Carbonation is a natural process where CO₂ 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 to react with CO₂.



The most reactive compounds for CO₂ 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 hydration 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, enriched in Mg, Fe and Ca silicates, are the main targets for *in situ* CO₂ mineralization projects, as discussed below.²¹

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 of 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 (Fig. 3).

Surface water flows through fractures originating a shallow aquifer open to the atmospheric CO₂ and O₂ fluxes; the water reacts with the peridotite and the pre-existing carbonate rocks in an open system becoming enriched in Mg²⁺ and HCO₃⁻. 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

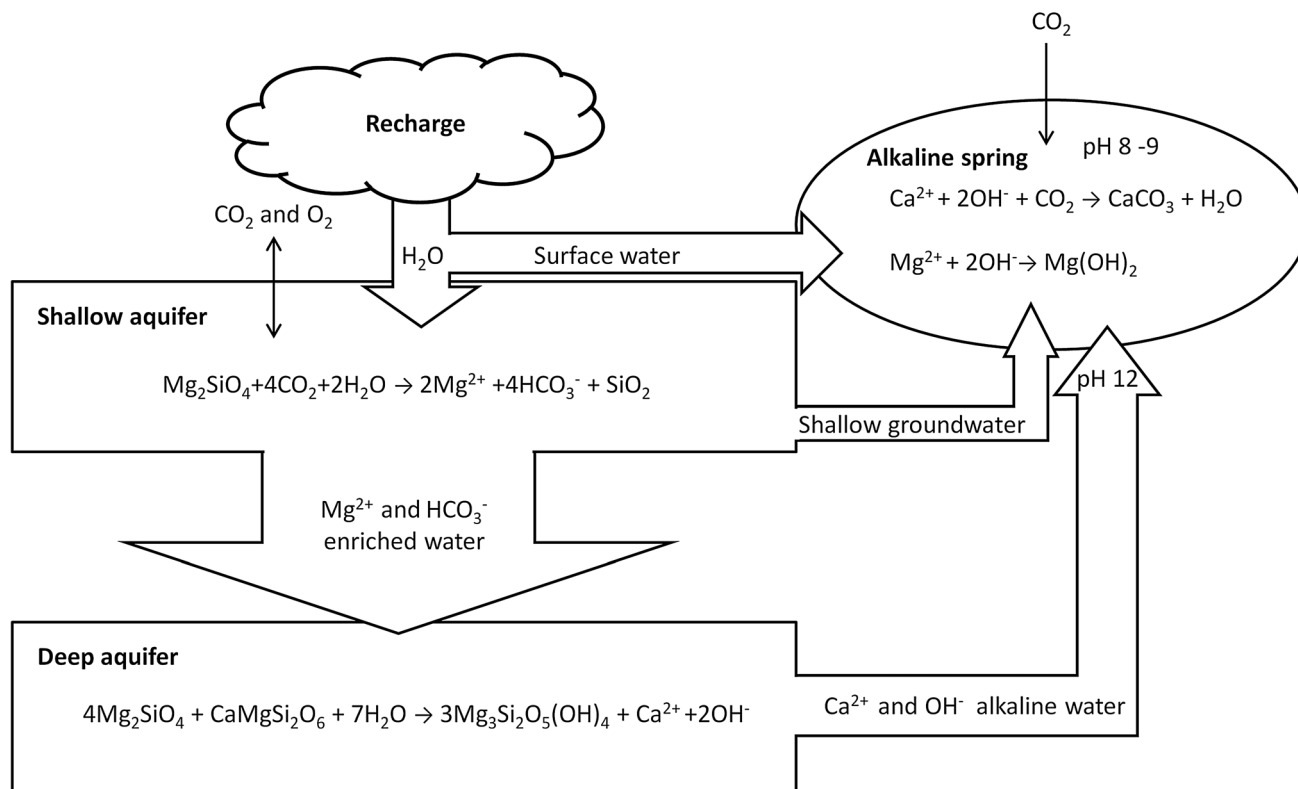


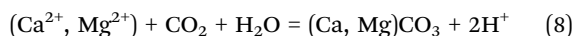
Fig. 3 Water fluxes and chemical reactions within peridotites.



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. The total volume of carbonate in the Samail Ophiolite is $5.5 \times 10^7 \text{ m}^3$ with an average age of 26 000 years indicating that about $4 \times 10^7 \text{ kg CO}_2$ per 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 5600 years for the deep reservoirs. Artificial enhancement of the carbonation can be achieved by 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 760 kJ kg⁻¹) and the geothermal gradient (up to 20 °C km⁻¹) 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}

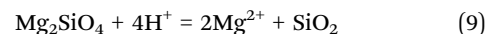
2.2 Basalts

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 water originating from aquifers within the hosting rocks at different depths and mineral concentrations. These aquifers are often enriched in ions including Ca²⁺ and Mg²⁺,³⁶ which can react with the injected CO₂ precipitating carbonates and releasing H⁺ as in reaction (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 reactions (9) and (10).



The availability of reactive Mg, Al and Ca silicates is therefore the controlling factor for the development of *in situ* CO₂ mineralization.³⁷

Following the injection of CO₂ (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; carbonate 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 CO₂ within the basalts of the ocean seafloor would benefit from a further series of trapping mechanisms in addition to the geochemical transformation of CO₂ in carbonates. The deep water environment, below 2700 m, and the cold temperature, below 2 °C, will make the injected CO₂ 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 CO₂ hydrates, where the CO₂ molecule is “encaged” within a lattice of ice strongly reducing its solubility in water in the case of seepage. Lastly, the thick sedimentary cover of the seafloor will form a low-permeability layer further reducing the possibility of leakage.³⁸

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

2.2.1 Onshore basalts. To validate the geochemical reactions during CO₂ 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 Palisades sill, composed of dolerite

Table 1 Location and characteristics of *in situ* basaltic carbonation areas

Location	Reservoir	Caprock	Storage potential	Ref.
New York state (USA)	Palisades sill. Dolerite with Ca-plagioclase and pyroxenes. Target zone porosity 5%	Lacustrine deposits of the Newark Basin; mudstone, arkoses, carbonate nodules shale and clastic sequences	An injection test was aimed to identify the buffering potential of dolerite	37–39
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%	Basalts layers with very low permeability	10 to 50 Gt CO ₂	40 and 41
Offshore USA East Coast	Sandy Hook Basin basalt porosity 15%	Sedimentary cover, mudstone, silt, clay	900 Mt 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% SiO ₂) basalt flows and hyaloclastite of olivine-tholeiite composition	Low-porosity basalts	12 Mt CO ₂	43–45



with Ca-plagioclase ($\text{CaAl}_2\text{Si}_2\text{O}_8$) and pyroxenes, and the underlying sedimentary rocks of the Newark Basin composed of 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 m^3 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 fluid injection. From these results it was estimated that 1000 m^3 of CO_2 saturated water at P_{CO_2} 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 $200\,000 \text{ km}^2$ 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 of over 300 different flows with a volume in excess of $224\,000 \text{ km}^3$. The flows can be divided into 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 through a well drilled in the Walla Walla County in south-eastern 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^{-1} and Fe 962 mg L^{-1} . In July 2013 a test injection of 1000 tCO_2 was started for one month. A model simulation forecasts that after one year 18% of CO_2 will be dissolved in groundwater within a radius of 55 m from the bore-hole. The storage capacity of the total basalt deposits for CO_2 *in situ* mineralization is estimated from 10 to 50 Gt 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 is evidence 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 of 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 incorporated 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 the CO_2 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 , and 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 of 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% SiO_2) 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 1500 years ago. The resulting structure is a sequence of more permeable lava flows hosting a shallow aquifer underlined by low-permeability layers of hyaloclastites separating and isolating a deeper aquifer. The shallow aquifer (above 400 m) has a 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, a pH of 8.4–9.4, CO_2 concentration below the atmospheric balance and it is depleted in O_2 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 km^3 ; CO_2 will be dissolved in water at P_{CO_2} of 25 bar, pH 3.7, DIC 1 mol kg^{-1} , requiring 22 t of water for each ton of CO_2 . A total of 2200 tCO_2 per year will be injected. A monitoring program including geochemical analysis, tracers and isotopic concentration measures will assess the diffusion of the injected CO_2 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 Mt CO_2 , 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 °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, the 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 of 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 water-rock interaction times and permitting chemical reaction with the surrounding altered basalt. The target for CO₂ injection is a volume of 7800 km³ of altered basalt with an average porosity of 10% for a storage potential of 920 Gt of CO₂.⁴⁸

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 an 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 CO₂ storage of about 1 km³ able to host 900 Mt of CO₂, trapping of which will be enhanced by long-term mineral carbonation reactions.⁵²

A general estimation of the overall potential storage capacity for ocean basalts is 8238 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 2700 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 CO₂ emission source, they have also been targeted as MC feedstock. The following sections will review the processes developed for both rocks and waste resources.

3.1 Processes developed for minerals

Since oxides and hydroxides of Ca and Mg are not abundant, silicate 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_{CO_2}). Serpentine, olivine, and to less extent wollastonite because of its lower abundance, are preferred based on performance and availability.⁵⁴

The sequestration of CO₂ 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

Table 2 Mineral chemistry, carbonation potential, and reactivity (modified from ref. 54). (Carbonation test conditions: 80% –37 μm feed; 1 hour; 185 °C; P_{CO_2} = 150 atm; 15% solids; 0.64 M NaHCO₃, 1 M NaCl). R_{real} = mass ratio of ore necessary to carbonate unit mass of CO₂; E_{CO_2} % = reaction efficiency, % stoichiometric conversion of Ca, Fe²⁺, and Mg cations in silicate feed to carbonate

Rock	Mineral	Mg	Ca	Fe ²⁺	R_{real}	E_{CO_2} (%)
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

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 CO₂ to form the desired carbonate.⁸

For each process carbonation route, key parameters of a process are presented in Table 3 to give quick insight into the main features. The parameters considered are the feedstock material used; the mineral to CO₂ 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 CO₂ (degree of conversion) (E_{CO_2} %). The different direct and indirect processes are discussed in detail in Section 3.1.2.

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 the direct aqueous mineral carbonation system, mainly due to the absence of protons at pH close to 7.^{61–63} In an 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 the 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, the precipitation of secondary phases decreases the dissolution



Table 3 Main characteristics and carbonation of minerals^{18,19,54,68,70,72,75–80,82,91,95,96,99–101,107–114,120–123,125}

Resource	R_{real}	Process conditions	E_{CO_2} (%)	Remarks
Serpentine	3.39	DCP: M, T (75 μm , 630 $^\circ\text{C}$), 115 bar, 155 $^\circ\text{C}$, 0.64 M NaHCO_3 , 1 M NaCl, 6 hours ^{54,82}	62	Cost of 210 \$ per tCO ₂ stored (lizardite)
Serpentine	5.25	DCP: M, T (75 μm :630 $^\circ\text{C}$), 115 bar, 155 $^\circ\text{C}$, 0.64 M NaHCO_3 , 1 M NaCl, 1 hour ^{95,96}	80	Cost of about A\$22 per tCO ₂
Serpentine	n.a.	DCP: M, T serp. (650 $^\circ\text{C}$), 30–45 bar, 140 $^\circ\text{C}$, 150–200 μm , NaHCO_3 ^{18,19,91}	n.a.	Flue gas is used
Serpentine	3.33	IC: M (37 μm), 120 bar, 300 $^\circ\text{C}$, NH_3 , 1 hour ¹²³	70	
Serpentine	2.33	DCP: M (100 μm), 50 bar, 550 $^\circ\text{C}$, weak acid/ NH_3 , 1 hour ¹²⁵	90	
Serpentine	3	IC: M (75 μm), 1 bar, 140 $^\circ\text{C}$ (300 $^\circ\text{C}$ regeneration), 1.4 M NH_4HSO_4 , NH_3 , 2–3 hours ^{68,72}	70	
Serpentine	5.38	IC: M (100 μm), 40 bar, 65 $^\circ\text{C}$, H_2SO_4 , NaOH, 3 hours ¹⁰⁹	55	
Serpentine	5.83	IC: M, T (250–425 μm), 20 bar, 500–550 $^\circ\text{C}$, $(\text{NH}_4)_2\text{SO}_4$, NH_3 , 0.5 hour ^{120–122}	60	Only 50–60% Mg is recovered
Serpentine	8.97	IC: M, T (50 μm :150), 1 bar, 325 $^\circ\text{C}$, HCl/NaOH, 26 hours ¹⁰⁸	26	
Serpentine	3.5	IC: M (<75 μm), 20 bar, 120 $^\circ\text{C}$, 0.1 M citrate, EDTA, 2 hours ^{110,111}	60	
Serpentine	3.23	IC: M (75 μm), 1 bar, 70 $^\circ\text{C}$, 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 \$ per tCO ₂ stored
Olivine	2.22	DCP: M (38 μm), 150 bar, 185 $^\circ\text{C}$, 0.64 M NaHCO_3 , 1 M NaCl, 3 hours ^{54,82}	68	
Olivine	1.82	DCP: M (<200 μm), 30 bar, 140 $^\circ\text{C}$, NaHCO_3 , 1–2 hours ^{18,19}	80–90	
Olivine	2.8	DC: M, C (10 μm), 1.2 bar, 500 $^\circ\text{C}$, bases, 0.5 hour ^{76,77}	25	A Spray Dry Absorber (SDA) is used
Olivine	n.a.	DC: M (2.5–60 μm), 500 $^\circ\text{C}$, 0.5 hour, moisture ⁷⁵	12	Flue gas is used dissolution of olivine in carbonated aqueous solutions using a flow-through column reactor
Olivine	n.a.	IC: M (75 μm), 250 $^\circ\text{C}$, 150 bar, 2 hours ¹⁰⁷	11	
Olivine	n.a.	DC: M (37 μm), 97 bar, 80 $^\circ\text{C}$, 20 hour ^{78,79}	8	
Wollastonite	n.d.	DCP: M (38 μm), 40 bar, 100 $^\circ\text{C}$, distilled water, 1 hour ^{54,82}	80	
Wollastonite	4.06	DC: M (38 μm), 20 bar, 200 $^\circ\text{C}$, 0.25 hour ^{70,80}	75	Cost of 102 euro per ton CO ₂ stored 90% dissolution was achieved in 1 hour, but carbonation was not tested
Wollastonite	n.a.	IC: 80 $^\circ\text{C}$, 30 bar, succinic acid ¹¹²	n.a.	
Brines	n.a.	DCP: 1 bar, 100 $^\circ\text{C}$, seawater/brine/alkalinity (NaOH or $\text{Mg}(\text{OH})_2$ or MgO) ^{99,100}	70	Amounts of NaOH and/or electricity
Brines	n.a.	DCP: 30 $^\circ\text{C}$, NaOH, NaCl, electrolysis ¹⁰¹	98	Potential co-removal of SO _x , NO ₂

M = mechanical pre-treatment; T = thermal pre-treatment; n.a. = not available; DC = direct carbonation; DCP = direct carbonation with pre-treatments; IC = indirect carbonation.

rates of those surfaces on which they precipitate; recent aqueous dissolution tests on diopside (25–70 $^\circ\text{C}$, in the presence of NaHCO_3) 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.

The particle size is a key parameter affecting carbonation because the reduction of size increases surface area and consequently the availability of reactive Mg and Ca.⁶⁵ The deposition of an inert layer such as SiO_2 on the surface limits the diffusion of the extraction solution into the particles.⁶⁶ 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 increases). 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 $^\circ\text{C}$ 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 the stirring rate (1500 rpm) can significantly influence the reaction rate in direct aqueous carbonation at the optimum temperature range (150–200 $^\circ\text{C}$). Operating above those values, carbonation was considered independent of the stirring rate and CO_2 pressure. The carbonation of wollastonite at constant temperature (150 $^\circ\text{C}$) 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 $^\circ\text{C}$, 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 the reaction rate of diluted CO_2 is slower because its dissolution rate is slower



compared to that of pure CO₂.⁷¹ The 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 CO₂ gas-liquid and solid-liquid mass transfer rates. Therefore, the lowest liquid-to-solid ratio in an autoclave reactor is 2 kg kg⁻¹, although the majority of aqueous carbonation experiments are carried out at a higher *L/S*-ratio to enhance the conversion efficiency.⁷² Also, the reduction of the *L/S*-ratio leads to a substantial improvement of the heat balance of the process and, thus, the overall CO₂ 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 CO₂ sequestration efficiency of the process.⁷⁰

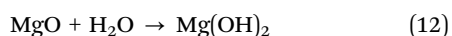
3.1.2 Direct carbonation (DC). DC consists of (1) gas-solid carbonation and (2) aqueous carbonation.

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:



High CO₂ pressures (100–150 bar) are necessary in order to obtain reasonable reaction rates.

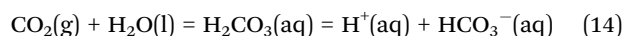
DaCosta and co-workers⁷⁵ developed a direct dry process for the sequestration of CO₂ where the flue gases pass through a bed of finely ground (2.5–60 μm) silicate rocks (mainly olivine, serpentine or wollastonite). As carbonation takes place, the mineral is replenished by either feeding fresh mineral with a pump or a conveyor. They reported that when using 5 g of olivine (surface of 2.5 m² g⁻¹) at temperature ranging from 100 to 500 °C and flue composition of 10% CO₂, 8.3% H₂O (balanced with N₂), the storage capacity was 0.12 g CO₂ per g olivine (12%) after 30 minutes.⁷⁵ A higher capacity of 18% was achieved capturing flue gas with 15% CO₂ in the presence of 8.3% water at 150 °C. The CO₂ stored decreased when 5% or 20% CO₂ gas stream were used, in the absence of moisture and at the higher temperatures tested, 175 and 200 °C.^{76,77} The enhanced CO₂ stored capacity in the 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 eqn (12) and (13):



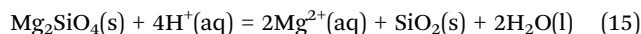
The above process is able to work in a dry environment, where moisture present in flue gas is assumed to be enough to convert the silicates oxides in the high reactive hydroxides and also requires only ~10–30 minutes, which represent a time scalable to the industrial level. However, it would require a

large amount of mineral per tonne of CO₂ sequestered due to the low efficiency (<20 g CO₂ per g olivine).⁷⁶ Based on the data available, more than 8 tonnes of olivine would sequester 1 tonne of CO₂. This would drastically reduce the applicability of this process to very small CO₂ emitters in terms of process size and material handling. Also, particle size reduction to <60 μm is very energy intensive.

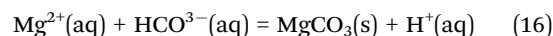
Aqueous carbonation: the carbonic acid route process involves CO₂ reacting at high pressure (100–159 bar) in an aqueous suspension with olivine or serpentine.^{61,69} Firstly, CO₂ dissolves in water and dissociates to bicarbonate and H⁺ resulting in a pH of about 5.0 to 5.5 at high CO₂ pressure:



Mg²⁺ is then liberated from the mineral matrix by H⁺:



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



Kwak and co-workers^{78,79} investigated the reaction pathways and reaction extent of direct aqueous carbonation of finely ground olivine (forsterite) (1 g) mixed with water (1 g) and fed into a batch reactor with a volume of 11.7 mL. The reaction was kept at 80 °C and 97 bar for 20 h with a final CO₂ storage capacity of 8%. The capacity was increased to 67% but it required 7 days.

Huijgen and co-workers⁸⁰ studied the direct aqueous carbonation of finely ground wollastonite mineral to particle size 38 μm 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 € per tCO₂ 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 € per tCO₂ sequestered, respectively.⁷⁰

Overall, direct routes present straightforward design and the absence of non-aqueous solvents. However, reaction conversions are low and high CO₂ pressure and temperature are required, compared to processes where pre-treatments are used to enhance the CO₂ storage capacity.⁷³ To enhance 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-pre-treatments have also been reported.



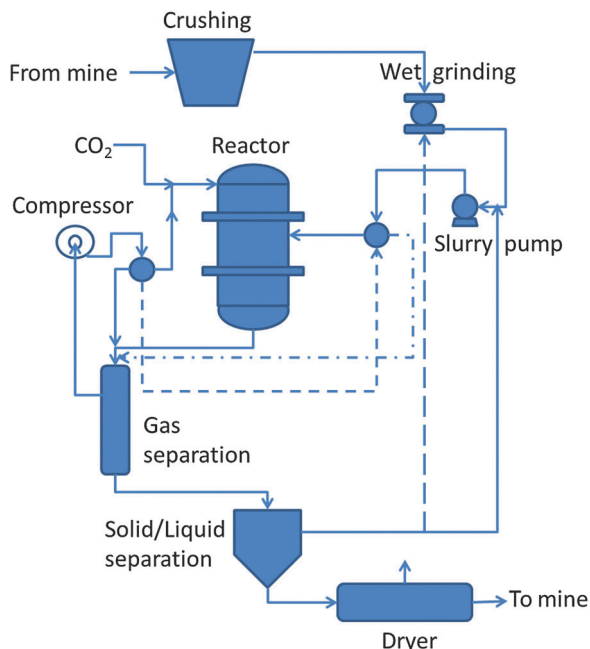


Fig. 4 Scheme of the NETL process (modified from ref. 82).

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 $<300\ \mu\text{m}$ 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 pebbles.⁸¹

The US National Energy Technology Laboratory (NETL) developed a direct carbonation process (scheme shown in Fig. 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 CaCO₃.⁸³

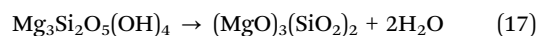
Various pre-treatment options such as ultrasonic treatment and wet grinding in caustic solution have also been tested, but they did not result in a higher reactivity.⁶¹ The major problem with many other pre-treatment options is the high energy input required.⁸⁴ Extensive studies on the mechanical activation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

silicates were performed at NETL^{85,86} and were reviewed later by Huijgen and Comans⁸⁴ 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 0.1 M 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.

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



The NETL findings indicate that the reaction rate for serpentine was slow if water (OH groups) was not removed. Thermally 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 kW h per 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 per t instead of 326 kW h per 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 per tCO₂ avoided,^{91,93} compared to \$ 210 per tCO₂ avoided in the NETL process.⁸² Balucan *et al.*⁹³ studied the thermal activation of serpentine from the Great Serpentine 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 20% residual hydroxyl groups, as opposed to the partly serpentinised ultramafic minerals of the Coolac Serpentine Belt. The activation strategy comprised heating serpentine




 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 (m ² g ⁻¹)	Crystalline phase (%)	Energy consumption (kW h per t)
As-received	—	—	0.25	100	—
Attritor	1500	10	7.3	45	190
		30	18.1	26	580
		120	35.2	12	2310
Planetary	450	10	4.8	37	2010
		30	5.2	17	6030
		10	3	57	640
Nutting	900	10	7.3	n.a.	170
		30	18.1	n.a.	520
		120	35.2	n.a.	2080
Thermal activation (serpentine)	Carbonation efficiency (%)		Energy consumption (kW h per t)		
Antigorite (75 μm)	62	—	—	—	293 (306)
Antigorite (38 μm)	92	—	—	—	293 (376)
Lizardite (75 μm)	9	—	—	—	326 (339)
Lizardite (38 μm)	40	—	—	—	326 (409)

particles with diameter < 34 μm 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 per t of serpentine.⁹³ Serpentine from Coolac Serpentinite Belt used in the experiments had a much larger particle size (D_{90} 127 μm) so the comparison is rather difficult. Also, particles size reduction to < 34 μm is energy intensive (> 220 kW h per 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 μm. 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 hydrocarbon-aceous fuelled furnace was used for the last heating step. The 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 per year of CO₂ emissions from a conventional pulverised fuel electricity generation plant (15 500 GW h per annum), which would consume about 41 Mt per year of serpentine and an additional 0.9 Mt per year coal to activate the serpentine at a claimed cost of about A\$22 per tCO₂.^{95,96} However, for the process neither experimental nor simulation work has 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 the long distance location of mineral and CO₂ point sources.

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 under 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 Fig. 5, namely Shell's direct pure and flue gas mineralization technology (Fig. 5a and b). Since flue gas with 10% vol CO₂ has a much lower solubility than pure CO₂ under pressure, leaching of cations in the presence of CO₂ will take place at a much slower rate. To avoid this, the flue gas is brought into contact with the mineral slurry prior to the 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 150–200 μm) for 1 hour at a temperature of 650 °C in a 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 the 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, resulting 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 is 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.

3.1.2.2.4 Brine-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 CO₂ from a

10 MW power generator at 90% efficiency for about 2 years.^{96,97} A diagram of the Calera process is depicted in Fig. 6. The technology foresees the introduction of brines containing alkaline earth metal ions into a reactor, where the brine is contacted with CO₂ containing gas. CO₂ dissolves in water to produce carbonate and bicarbonate ions, resulting in a decrease in the pH of the solution. In order to produce a 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 increased 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 to be 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 with 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 a significant scale) reduces the wide application of this technology.⁹⁸

Fig. 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

Another method of sequestering carbon dioxide using brines, referred to as SkyMine, has been recently proposed.¹⁰¹ CO₂ is absorbed into an aqueous caustic soda mixture to form carbonate and/or bicarbonate products.¹⁰¹ Flue gas from the power plant is cooled from 300 °C to 30 °C in a series of heat exchangers and then it is introduced at the bottom of an

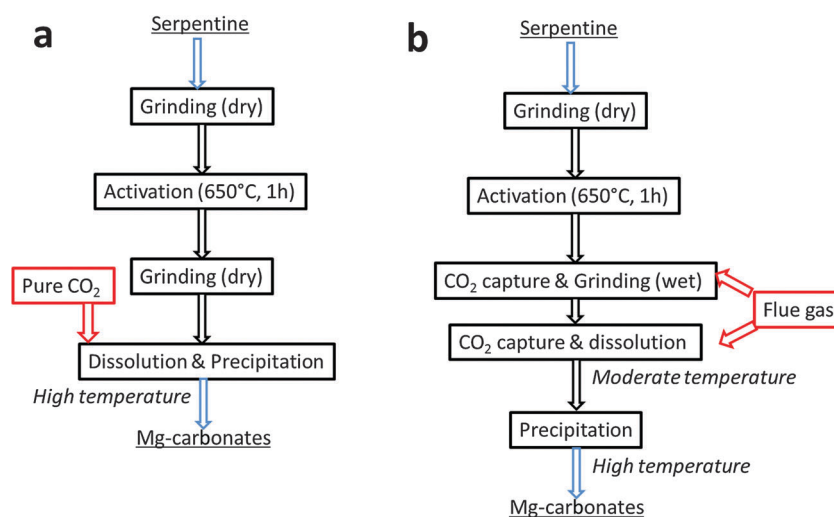


Fig. 5 Mineralization process concept for pure CO₂ and flue gas (modified from ref. 19).



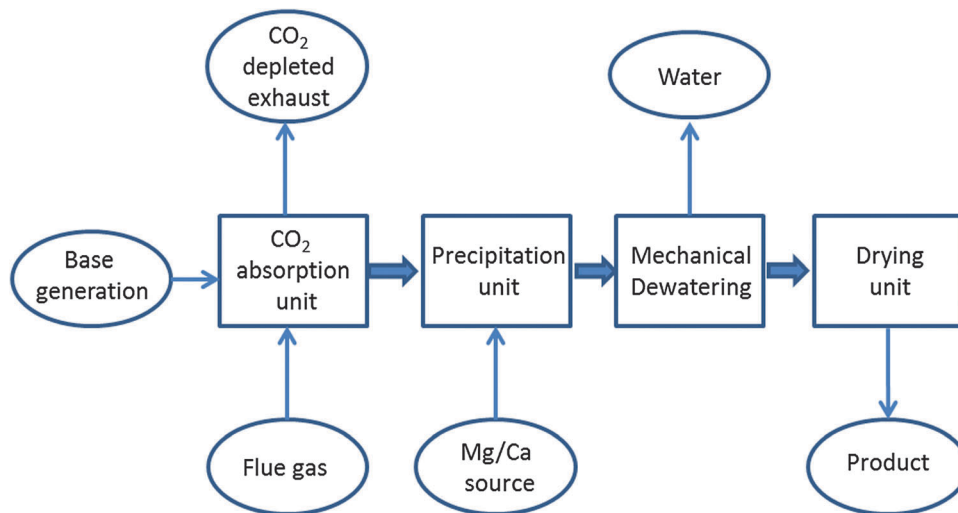
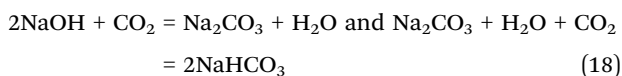
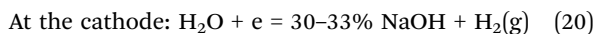
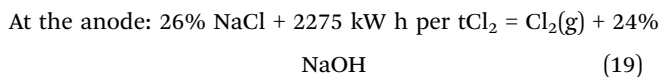


Fig. 6 Calera process diagram (modified from ref. 98).

absorber, where NaOH is used. The latter is produced by brine electrolysis. The reaction taking place in the carbonation chamber is:



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 Na₂CO₃ is achieved) or heat derived from hydrogen produced in electrolysis. In the membrane cell processing units, the following inputs and products are obtained:



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 SO_x and NO₂ from the flue gas, and remove heavy metals, such as mercury. A joint venture namely Skyonic Corporation, which includes BP and Conoco-Phillips, has started the construction of a commercial CO₂ capture plant to remove 83 000 tCO₂ 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 publically available. Also, it has to be noticed that market for HCl and sodium bicarbonate is 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 increase 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 than water in accelerating silicate leaching dissociation of Mg–O–Si bonds in the presence of protons.^{59,105} Bonfils work showed that the presence of disodium oxalate under 20 bar of CO₂ pressure leads 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, salicylate and humate. Their study, in agreement with previous reports, concluded that aqueous organic ligands have at most a small effect on forsterite dissolution rates



under 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 were related to their aqueous speciation, since these organic species are in the form of neutral species at acidic pH, but as negatively charged aqueous species under 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 the 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 μm 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% was obtained after mechanical pre-treatment (211 kW h per t) after 2 and 18 hours, respectively.⁶⁰ However, as high carbonation efficiency was only obtained after long times, the process is not viable on 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 $\text{Mg}(\text{OH})_2$ 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 H_2SO_4 at a temperature 20–65 °C for 3–12 hours. The resulting magnesium sulphate was reacted with sodium hydroxide to precipitate $\text{Mg}(\text{OH})_2$ following an exothermic reaction. $\text{Mg}(\text{OH})_2$ subsequently reacted with CO_2 in aqueous suspension at 20 °C and 40 bar. A conversion of 55% was achieved in 10 minutes under these mild conditions. Sulphuric acid was regenerated by reacting CO_2 with MgSO_4 .¹⁰⁹ However, chemicals make up and intensive chemical regeneration hindered the deployment of this process. The effect of HCl, H_2SO_4 and HNO_3 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: $\text{H}_2\text{SO}_4 > \text{HCl} > \text{HNO}_3$. 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 with strong acids. Teir *et al.*⁵⁶ found that acetic acid

(CH_3COOH) and formic acid (HCOOH) were able to leach a significant amount of magnesium from serpentine. Krevor and Lackner¹¹⁰ 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 75 μm was reacted in a solution with dissolved salts under a CO_2 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 strong 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 the SiO_2 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 by 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 of improving the efficiency of mineral dissolution and recovering and re-using additives, Maroto-Valer and co-workers proposed a pH-swing CO_2 mineralisation process using ammonium salts.⁶⁸ At 100 °C, 1.4 M aqueous solution NH_4HSO_4 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 Fig. 7. In the first step, NH_3 was used to capture CO_2 from flue gas to produce NH_4HCO_3 . In the mineral dissolution step, 1.4 M NH_4HSO_4 was used to extract Mg from serpentine ground to a particle size range 75–125 μm . The Mg-rich solution was then neutralised by adding NH_4OH , after which impurities in the leaching solution were removed by adding NH_4OH . The Mg-rich solution is then reacted with the product from the capture step NH_4HCO_3 to precipitate carbonates. Since the formation and stability of hydro-carbonates is temperature dependent, $\text{MgCO}_3 \cdot 3\text{H}_2\text{O}$ (nesquehonite) can be converted to $4\text{MgCO}_3 \cdot \text{Mg}(\text{OH})_2 \cdot 4\text{H}_2\text{O}$ (hydromagnesite) at temperatures above 70 °C. Precipitation of hydromagnesite resulted in a solution mainly containing $(\text{NH}_4)_2\text{SO}_4$. The final step was the additive regeneration, with the decomposition of $(\text{NH}_4)_2\text{SO}_4$ at ~ 330 °C, and producing



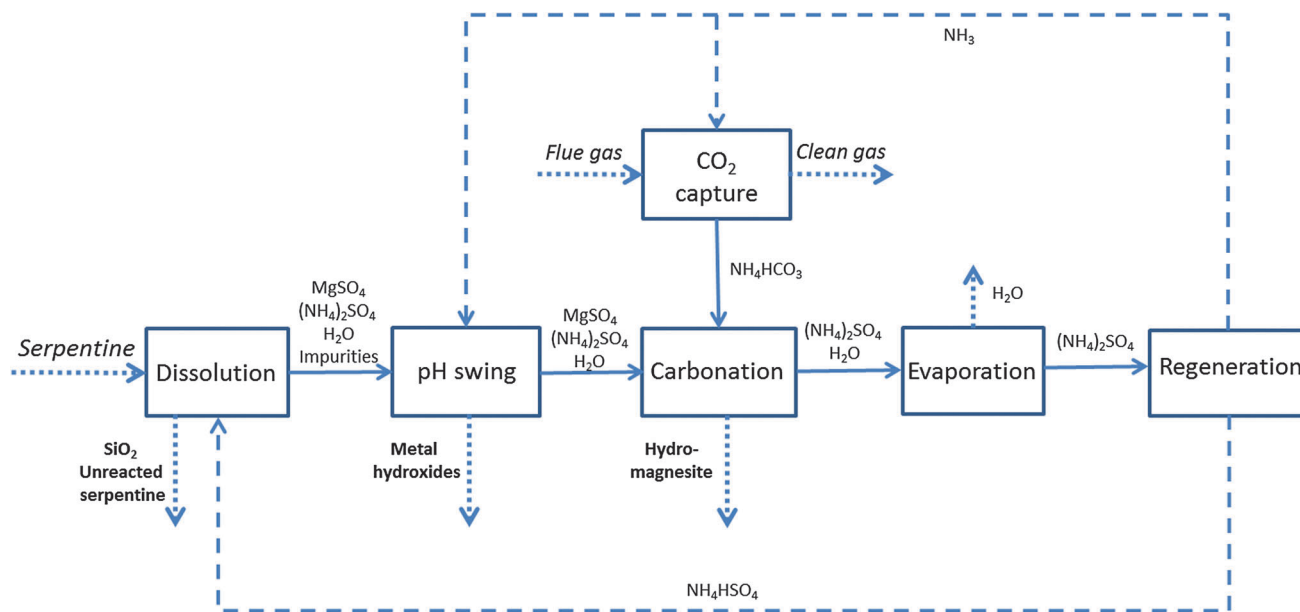


Fig. 7 pH-swing CO_2 mineral carbonation process with recyclable ammonium salts (modified from ref. 68).

NH_3 for the capture step and NH_4HSO_4 for the dissolution step.^{68,72}

In a typical capture process, CO_2 is first absorbed by chemicals (*e.g.* NH_3) 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 CO_2 captured as sodium carbonate/bicarbonate is directly used in the proposed mineral carbonation, there is no need for desorption and compression of CO_2 . 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 CO_2 capture process, which has been demonstrated to capture more than 90% of CO_2 (from 3–15% CO_2 in flue gas)¹¹⁵ and an estimated energy penalty of 477 kW h per tCO_2 .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 of 50 g L^{-1} was used, 50–56 tH_2O were required to sequester 1 tCO_2 . 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 300 g L^{-1} , 16 tH_2O were required to sequester 1 tCO_2 . However, since the CO_2 fixation efficiency decreased to 46.6%, a 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 Fig. 8. In the first step, $\text{Mg}(\text{OH})_2$ was produced from serpentinite and in the second step, $\text{Mg}(\text{OH})_2$ was carbonated in the dry phase. This process takes advantage of: (1) the higher reactivity of $\text{Mg}(\text{OH})_2$ compared to that of serpentinite and MgO and (2) the potential recovery of the heat of reaction released during the carbonation.

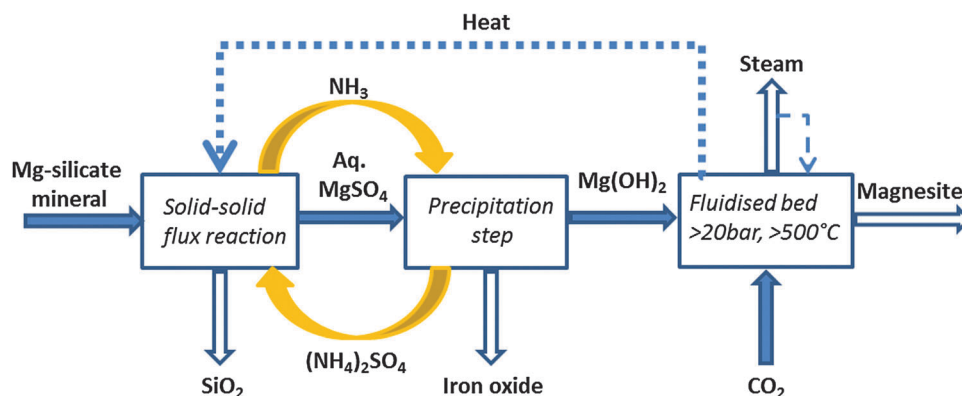


Fig. 8 Åbo Akademi mineral carbonation process (modified from ref. 121).



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 magnesium sulphate, which was then dissolved in water. Adjusting the pH using ammonium hydroxide or ammonia led to precipitation of Mg(OH)₂, while iron oxide was recovered as the 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 recovery 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}

Hunwick¹²³ 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 (<40 μm) Mg/Ca silicates were mixed with ammonia to produce an aqueous slurry of 30% solids. Then, CO₂ 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 100 m 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 from/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 minerals¹²⁷ 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 be 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 material¹²⁸ as well as pure and precipitated Ca or Mg-carbonates.^{126,129} On the other hand, the amount of industrial waste materials available is 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, the gas flow rate, the liquid flow rate, particle size, and solid pretreatment.^{24,128,130,131}

The theoretical maximum CO₂ uptake (tCO₂ uptake, eqn (21)) of waste expressed in wt% was calculated using a modified Steinour formula;^{25,132} E_{CO_2} indicates the experimental CO₂ uptake.

$$\begin{aligned} \text{tCO}_2 \text{ uptake} = & 0.785 \times (\% \text{CaO} - 0.53 \times \% \text{CaCO}_3 - 0.7 \times \% \text{SO}_3) \\ & + 1.091 \times \% \text{MgO} + 0.71 \times \% \text{Na}_2\text{O} + 0.468 \\ & \times (\% \text{K}_2\text{O} - 0.632 \times \% \text{KCl}) \end{aligned} \quad (21)$$

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; the % of CaO and MgO, the theoretical and experimental CO₂ capture capacity and the different process conditions were investigated.

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 (eqn (22)) is followed by simultaneous dissolution of Ca(OH)₂ (eqn (23)) and dissociation of aqueous CO₂ (eqn (24) and (25)) precedes the carbonation reaction (eqn (25)).^{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

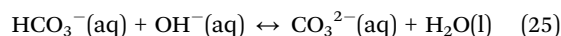
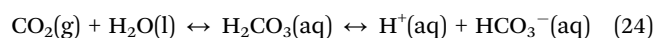
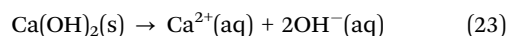
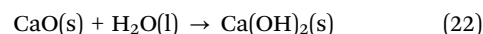


Table 5 Main properties and carbonation conversions of waste materials (DC – direct carbonation, DAC – direct aqueous carbonation, IAC – indirect aqueous carbonation, NW – natural weathering, CRW – cold rolling wastewater, DI – deionized water)

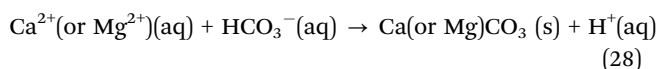
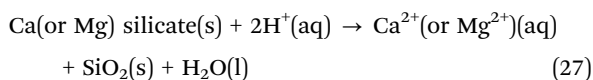
	CaO (%)	MgO (%)	tCO ₂ (%)	Process	E _{CO₂} (%)	Remarks	
Iron and steelmaking slags							
BFS ^{24,25,133}	15–42	5–11	20–44 ²⁵	IAC: (1) step ($T = 70\text{ }^{\circ}\text{C}$; CH ₃ COOH); (2) step ($T = 30\text{ }^{\circ}\text{C}$; $P = 1\text{ bar}$; NaOH) ¹³⁴ DAC: $P = 5\text{ bar}$; $L/S = 0.15$; 100% CO ₂ ; $t = 2\text{ h}$ ¹³⁷	22.7 7	+High CO ₂ sequestration capacity; +generated in large quantities; +generated near CO ₂ source; +carbonation improves mechanical and environmental parameters of slag; +possible applications: PCC; –must undergo milling, except AODS and LFS; –high T&P, US (€4000 per t slag ¹³⁵), or additives are needed for acceptable conversion; cost: €77 per t-CO ₂ net avoided (DAC) ⁷⁰ ; IAC with PCC production: 300 kW h per t-CO ₂ ⁶⁶ or €1990 per t-CaCO ₃ for chemicals only ¹³⁶	
BOFS ^{25,133,138}	34–56	2–6	29–52 ²⁵	DAC: $T = 100\text{ }^{\circ}\text{C}$; $P = 19\text{ bar}$; $L(\text{DI})/S = 2$; 25 100% CO ₂ ; $t = 30\text{ min}$; $d < 38\text{ }\mu\text{m}$ ¹³⁹ DAC: $T = 70\text{ }^{\circ}\text{C}$, $P = 1\text{ bar}$; $L(\text{DI})/S = 10$; 22.9 100% CO ₂ ; $t = 2\text{ h}$ ¹³¹ DAC: $T = 60\text{ }^{\circ}\text{C}$; $P = 1.47\text{ bar}$; $L(\text{DI})/S = 20$; 100% CO ₂ ; $t = 30\text{ min}$ ¹⁴⁰ DAC: $T = 25\text{ }^{\circ}\text{C}$; $P = 1\text{ bar}$; $L(\text{CRW})/S = 20$; $t = 125\text{ min}$; $d < 44\text{ }\mu\text{m}$ ¹⁴¹ IAC: (1) step ($T = 80\text{ }^{\circ}\text{C}$, NH ₄ Cl; ground $d < 2000\text{ }\mu\text{m}$); (2) step (13% CO ₂) ⁶⁶ IAC: (1) step ($T = 30\text{ }^{\circ}\text{C}$; CH ₃ COOH); (2) step ($T = 30\text{ }^{\circ}\text{C}$; $P = 1\text{ bar}$; NaOH; 100% CO ₂) ¹²⁶	28.9 28.3 16.2 9		
EAFS ^{25,133,138,142}	25–47	4–19	24–48 ²⁵	DAC: $T = 20\text{ }^{\circ}\text{C}$; $P = 1\text{ bar}$; $L(\text{DI})/S = 10$; 15% CO ₂ ; $t = 40\text{ h}$; $d = 38\text{--}106\text{ }\mu\text{m}$ ¹⁴² DAC: $T = 20\text{ }^{\circ}\text{C}$; $P = 1\text{ bar}$; $L(\text{DI})/S = 10$; 15% CO ₂ ; $t = 65\text{ min}$; $d < 100\text{ }\mu\text{m}$ ¹⁴³ DAC: $P = 5\text{ bar}$; $L/S = 0.15$; 100% CO ₂ ; $t = 2\text{ h}$ ¹³⁷	1.74 1.9–8.7 12		
LFS ^{142,143}	42–58	6–15	42 ¹⁴²	DAC: $T = 50\text{ }^{\circ}\text{C}$; $P = 3\text{ bar}$; $L/S = 0.4$ ¹⁴⁴ DAC: $T = 20\text{ }^{\circ}\text{C}$; $P = 1\text{ bar}$; $L(\text{DI})/S = 10$; 15% CO ₂ ; $t = 40\text{ h}$; $d = 38\text{--}106\text{ }\mu\text{m}$ ¹⁴² DAC: $T = 20\text{ }^{\circ}\text{C}$; $P = 1\text{ bar}$; $L(\text{DI})/S = 10$; 15% CO ₂ ; $t = 65\text{ min}$ ¹⁴³	18 24.7 4.6		
AODS ^{133,145}	41–61	4–7.5	31–54 ^{25,135}	DAC: $T = 50\text{ }^{\circ}\text{C}$; US; $L(\text{DI})/S = 10$; 100% CO ₂ ; $t = 4\text{ h}$; $d = 63\text{--}200\text{ }\mu\text{m}$ ¹³⁵ DAC: $T = 50\text{ }^{\circ}\text{C}$; US; $L(16.6\text{ g L}^{-1}\text{ MgCl})/S = 10$; 10% CO ₂ ; $t = 240\text{ min}$ ¹⁴⁵ DAC: $T = 90\text{ }^{\circ}\text{C}$; $P = 9\text{ bar}$; $L/S = 16$; $t = 120\text{ min}$ ¹⁴⁶	15.1 27 26.4		
Cement wastes							
CKD ^{25,132,148,149}	34–48	1–1.5	10–30	DAC: ambient T & P ; $L/S = 0.33$; 5–15% CO ₂ ; 8 h ¹³² DAC: ambient T , $P = 2\text{ bar}$; H ₂ O to form paste; 100% CO ₂ ; $t = 72\text{ h}$ ¹⁴⁹	8–18 10		+Generated in large quantities; +generated near CO ₂ source (CKD, CBD); +carbonated product can be reused in cement manufacturing, aggregates, to produce PCC <i>etc.</i> ; +CKD, CBD has fine particle size; –waste cement needs to be ground; –low carbon sequestration capacities (CKD); cost: IAC with PCC production: \$136–323 per t-CaCO ₃ ¹⁴⁷
CBD ^{148,149}	66	1	50	DAC: ambient T , $P = 2\text{ bar}$; H ₂ O to form paste; 100% CO ₂ ; $t = 72\text{ h}$ ¹⁴⁹	25		
Waste cement, RCA ^{150,151}	25–63	0.3–2	20	DAC: $T = 20\text{ }^{\circ}\text{C}$; $P = 1\text{--}4\text{ bar}$; $L/S = 0.25\text{--}0.5$; 0.03–100% CO ₂ ; 0.8–100 h; $d < 1.8\text{ mm}$ ¹⁵⁰ DAC: $T = 20\text{ }^{\circ}\text{C}$; $L(\text{DI})/\text{cement} = 0.26$; 20% CO ₂ ; $t = 60\text{ min}$ ¹⁵¹ IAC: (1) step ($T = 50\text{ }^{\circ}\text{C}$; $P = 30\text{ bar}$; 100% CO ₂ $L/\text{cement} = 350$; $d = 10\text{--}200\text{ }\mu\text{m}$; $t = 10\text{ min}$). (2) Step ($T = 30\text{ }^{\circ}\text{C}$; 1 bar; $t = 30\text{ min}$) ¹⁴⁷	1.6–16.5 8.9		
MSWI ashes							
MSWI BA ^{25,130,149}	22–53	2.8	25	DAC: $P = 3\text{ bar}$; $L/S = 0.3\text{--}0.34$; 100% CO ₂ ; RH = 65%; $t = 2.5\text{ h}$; $d < 710\text{ }\mu\text{m}$ ¹⁵² DAC: ambient T , $P = 2\text{ bar}$; H ₂ O to form paste; 100% CO ₂ ; $t = 72\text{ h}$ ¹⁴⁹	3.2	+Produced in large quantities (MSWI BA); +produced near CO ₂ source; +carbonation reduces pH and leaching of hazardous elements for safer landfill; +grinding not required; –low carbon sequestration capacity (MSWI BA)	
APC residue ^{25,152,153}	36–60	1–2.5	50–58	DAC: $P = 3\text{ bar}$; $L/S = 0.2\text{--}0.3$; 100% CO ₂ ; RH = 65%; $t = 2.5\text{ h}$; $d < 212\text{ }\mu\text{m}$ ¹⁵² DAC: $T = 20\text{--}30\text{ }^{\circ}\text{C}$; $L/S = 0.3$; 20% CO ₂ ; $t = 50\text{--}150\text{ min}$; $d_{\text{mean}} = 66\text{ }\mu\text{m}$ ¹⁵³ DC: $T = 650\text{--}500\text{ }^{\circ}\text{C}$; $P = 1\text{ bar}$, 10–50% CO ₂ ⁶⁷ $T = 30\text{--}50\text{ }^{\circ}\text{C}$; $P = 1\text{--}10\text{ bar}$, 100% CO ₂ , $L/S = 0\text{--}0.6$ ⁶⁷	7.3 8–12 25 25		



Table 5 (continued)

	CaO (%)	MgO (%)	tCO ₂ (%)	Process	E _{CO₂} (%)	Remarks
Fuel combustion ashes						
Coal FA ^{25,154,155}	1.3–10	1–3	6–9	DAC: <i>T</i> = 20–60 °C; <i>P</i> = 10–40 bar; 100% CO ₂ ; <i>L/S</i> = 10; <i>d</i> _{median} = 40 μm; 18 h ¹⁵⁴ DAC: <i>T</i> = 30 °C; 90 °C; <i>P</i> = 10–40 bar; <i>L/S</i> = 1–10; 100% CO ₂ ; ground <i>d</i> = 20–150 μm or <150 μm ¹⁵⁵ DAC: <i>T</i> = 90 °C; <i>P</i> = 40 bar; <i>L/S</i> = 1; bulk ash; <i>t</i> = 2 h ¹⁵⁷ NW: ambient <i>T&P</i> ; wet deposited ash; <i>t</i> = 20 year ¹⁵⁷	2.6 3.6–7.2 6.5 6.8	+Produced in large quantities (coal FA); +produced near CO ₂ source; +grinding usually not required; +high CO ₂ sequestration capacity (OS FA); –low CO ₂ sequestration capacity (coal FA); –waste available in few areas (OS FA); cost: \$11–21 per t-CO ₂ at mineralization capacity of 0.1–0.2 tCO ₂ per t-FA. ¹⁵⁶
Lignite FA	27.5	6.5	43	DAC: <i>T</i> = 75 °C, <i>P</i> = 1 bar; 10% CO ₂ ; <i>L/S</i> = 20; 4.5 h; <i>d</i> < 250 μm ¹⁵⁸ DAC: <i>T</i> = 30–80 °C; <i>L/S</i> = 40–80; NaCl 1–25 g L ⁻¹ ; pH = 5–9; 100% CO ₂ ; ground <i>d</i> = 30–125 μm; <i>t</i> = 10–50 min ¹⁵⁹ DAC: ambient <i>T&P</i> ; <i>L/S</i> = 10; 15% CO ₂ ; <i>t</i> = 65 min ¹⁴³ NW: ambient <i>T&P</i> ; wet deposited ash; <i>t</i> = 8 w ¹⁶¹	23 7.1 29 2.2	
OS FA (PF, CFB) ^{160,161}	38–50	5–12	26–49			
WA ¹⁴⁹	24–46	8–9	50	DAC: ambient <i>T</i> , <i>P</i> = 2 bar; H ₂ O to form paste; 100% CO ₂ ; <i>t</i> = 72 h ¹⁴⁹	8	
Mine tailings						
Asbestos tailings ^{23,164}	0.2	39	43	DC: <i>T</i> = 375 °C; <i>P</i> = 1 bar; 56% CO ₂ ; 10% H ₂ O; <i>d</i> = 37–75 μm; <i>t</i> = 5 h ¹⁶⁴	0.5	+Carbonation destroys the asbestos nature (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 E _{CO₂} ; –too expensive to achieve high carbonate conversion (asbestos and Ni tailings); –bicarbonates generated instead of carbonates (RM); cost: \$147 per t-CO ₂ for DAC of RM; ¹⁶² IAC of Ni tailings with hydromagnesite production US\$600–1600 per t-CO ₂ for chemicals only ¹⁶³
Ni tailings ^{23,165}	3.4	21–40	43	IAC: (1) step (<i>T</i> = 70 °C; <i>L</i> (4 M HCl, HNO ₃)/ <i>S</i> = 10; <i>t</i> = 2 h; ground <i>d</i> < 0.5 mm); (2) step (<i>T</i> = 30 °C; 100% CO ₂ ; NaOH; <i>t</i> = 0.5 h) ¹⁶⁵	29	
RM ^{23,162,166}	2–7	<1	7–19	DAC: <i>T</i> = 20 °C; <i>P</i> = 3.5 bar; <i>L/S</i> = 0.2–0.6; ground <i>d</i> _{mean} = 30 μm; <i>t</i> = 12 h ¹⁶⁷ DAC: ambient <i>T&P</i> ; 100% CO ₂ ; ground <i>d</i> = 0.1–160 μm; 3 carbonation cycles (each 5 h) ¹⁶² DAC: ambient <i>T&P</i> ; <i>L/S</i> = 10; 15% CO ₂ ; <i>d</i> < 1000 μm; <i>t</i> = 24 h ¹⁶⁶	5.3 7.2 4.15	
Alkaline paper mill wastes						
APMWA ^{148,149,168}	45–82	1–5	42–55	DAC: <i>T</i> = 30 °C; <i>P</i> = 10 bar; <i>L</i> (DI)/ <i>S</i> = 20; <i>d</i> = 15 μm; <i>t</i> = 2 h ¹⁶⁸ DAC: ambient <i>T</i> , <i>P</i> = 2 bar; H ₂ O to form paste; 100% CO ₂ ; <i>t</i> = 72 h ¹⁴⁹	21.8 10–26	+High carbon sequestration capacity; +grinding not required; –generated in small quantities

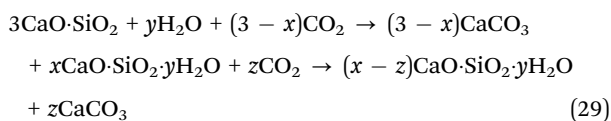
The aqueous carbonation of wastes, in which CaO is bound as a silicate (such as steel slags *etc.*), can in general be expressed using eqn (24), (27) and (28).^{21,139,170} Firstly CO₂ dissolves in the aqueous phase resulting H⁺-ions (eqn (24)). Secondly, Ca (Mg) leaches from the mineral matrix due to a slightly acidic environment (eqn (27)). Finally, Ca (Mg) carbonate precipitates (eqn (28)). 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}



It has also been demonstrated that contaminated solids of cementitious nature can be rapidly remediated whilst binding

CO₂ 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) CO₂ diffusion in air and (2) permeation through the solid is followed by (3) solvation of CO₂(g) to CO₂(aq), (4) hydration of CO₂(aq) to H₂CO₃, (5) ionization of H₂CO₃ to H⁺, HCO₃³⁻ and CO₃²⁻, (6) dissolution of cementitious phases (Ca₃SiO₅, Ca₂SiO₄) releasing Ca²⁺ and SiO₄²⁻ 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 (eqn (29)).^{24,172} The extent and rate of carbonation depend mainly on the diffusivity and reactivity of CO₂, which in turn depend on the binder type and hydration degree as well as pore type and process conditions (CO₂ partial pressure, relative humidity, temperature and pressure).¹⁷²





Generally, the waste carbonation reaction could occur in four routes: (1) conversion inside the solid particle, (2) CaCO_3 crystallization on the 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 $\text{Ca}(\text{OH})_2$ 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 the rate-limiting mechanism.^{132,174–176}

In the next section, 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 per t-steel¹⁴²), accounting for 6–7% of global CO_2 emissions.¹³⁸ Also, globally, these processes generate about 315–420 Mt per year.¹²⁶

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 CO_2 emissions are estimated to be up to 171 Mt of CO_2 ,¹³⁴ representing about 0.6% of global CO_2 emissions from fuel combustion.²³ In general steel-making slags require grinding as carbonation pre-treatment,^{23,48} but the cost of mining and transportation to CO_2 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 CO_2 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 (eqn (27)) after rapid carbonation of free CaO (eqn (22)–(26)) and then carbonated (eqn (28)) and on the particles' surface. Ca diffusion through the solid matrix was the rate-limiting step due to the formation of a

CaCO_3 capsule and Ca-depleted silicate zone.^{23,139,176} The E_{CO_2} 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 per t-slag).¹³⁵ Particle size was also an important variable, as carbonation was significantly improved by using smaller fractions (38–106 μm).^{25,139,142,176} Estimations have shown that cost of slag (200 °C and 20 bar of pure CO_2) was €77 per t- CO_2 net avoided.⁷⁰ Carbonation of Ca-carrying cementitious materials to sequester CO_2 also resulted in the development of high early stage strength for building materials applications, achieving CO_2 uptake of 7–12% in the process.¹³⁷ A possibility to upgrade steel slags into products of high commercial value, such as high-purity precipitated CaCO_3 (PCC), has also been addressed in several studies.^{66,126,133} A number of extraction agents including HNO_3 , H_2SO_4 , NaOH ,¹³⁸ NH_4Cl ,^{66,106} and CH_3COOH ,^{126,133} $\text{CH}_3\text{COONH}_4$, NH_4NO_3 ¹⁰⁶ NH_4HSO_4 ¹⁷⁹ have been investigated for the indirect carbonation route. The use of HNO_3 solution resulted in rapid extraction of Ca and Mg from BOF and EAF slags with CO_2 sequestration capacity of 0.26–0.38 t- CO_2 per t-slag.¹³⁸ In the case of NH_4Cl (at 80 °C), 60% of Ca was extracted resulting in PCC of 98% purity. The CO_2 sequestration capacity of 16% (0.16 t- CO_2 per t-slag) was achieved in the process with a total energy consumption of 300 kWh per t- CO_2 .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 the acetic acid route would cost €1990 per t- CaCO_3 assuming that the byproduct sodium acetate could be sold for ~€680 per t.¹³⁶ A closed loop multi-step process was developed to extract Ca^{2+} from steel slag with NH_4HSO_4 solution to form solid CaSO_4 , which after pH adjustment and precipitation of impurities reacted with $(\text{NH}_4)_2\text{CO}_3$ (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 CO_2 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 (SiO_2 , $\text{Ca}_2\text{Al}[\text{AlSiO}_7]$), sulfates (CaSO_4 , $\text{CaSO}_4\cdot 2\text{H}_2\text{O}$, $\text{Ca}_6\text{Al}_2(\text{SO}_4)_3(\text{OH})_{12}\cdot 26\text{H}_2\text{O}$), carbonates (CaCO_3), but also metal oxides, hydroxides ($\text{Ca}(\text{OH})_2$, Fe_2O_3 , Fe_3O_4) and chlorides.^{25,130} As Ca and Mg contents of MSWI BA are typically too low for significant CO_2 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, CO₂ partial pressure, liquid to solid ratio (optimum $L/S \sim 0.3$ w/w^{152,182}), residence time, and particle size^{152,183} have also been investigated. MSWI BA has shown CO₂ 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 CO₂ uptake 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 CO₂ binding reactions. The dissolution of Ca from the solid matrix into the liquid phase and the diffusion of CO₂ into the pores have been reported as the rate-limiting steps.¹³⁰ The kinetics of CO₂ uptake include the two following reaction steps: (1) an initial rapid CO₂ 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 CO₂ 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 a similar trend as for MSWI BA, only with higher weight gains attributed to higher lime contents and larger specific surface areas.¹⁵² The E_{CO_2} uptake of the APC residues ranges from 7 to 25 wt% (0.07–0.25 t-CO₂ per t-APC)^{25,67,152,153,188} (Table 5). Baciocchi *et al.* compared dry (1 bar, 10–50% CO₂, 350–500 °C) and wet routes (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 a potential CO₂ storage capacity of 0.25 t per t-APC residue, 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 μm), whose composition varies depending on the mineral content of fuel. Generally, it consists of an amorphous aluminosilicate glass matrix (Si_xAl_yO_z) and

recrystallized minerals, including quartz (SiO₂), cristobalite (SiO₂) and mullite (3Al₂O₃·2SiO₂).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 CO₂ 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 the 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 a 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. A preliminary economic analysis of the process reported that 90% CO₂ capture from a 532 MW power plant would cost about \$11–21 per t-CO₂ assuming a sequestration capacity of 0.1–0.2 t-CO₂ per 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 a significant amount of CKD is already carbonated (CKD contains 46–57% CaCO₃¹³²). 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 per t-CBD).¹⁴⁸ The CKD E_{CO_2} uptake achieved was 8–25 wt% (*i.e.* 0.08–0.25 t-CO₂ per t-CKD,CBD) at ambient temperature and pressure in a column reactor¹³² 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 is eliminated.¹²⁸

Waste cement is a by-product obtained from the 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 CO₂ 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 E_{CO_2} 16.5%, at 100% CO₂ and 4 bar, Table 5). Kashef-Haghighi and Ghoshal¹⁵¹ achieved a carbonation efficiency of 18% and an E_{CO_2} of 8.9% by curing fresh concrete blocks in a flow-through reactor (20% CO₂ in N₂, 20 °C and 60 min). A small demonstration scale is planned for the 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.¹⁹²

Katsuyama *et al.*¹⁴⁷ and Iizuka *et al.*¹⁷¹ produced high purity CaCO₃ using an indirect aqueous carbonation route for the extraction of Ca²⁺ from cement waste by pressurized CO₂ (30 bar) and subsequent carbonation at reduced pressures (1 bar). The estimated costs per 1 metric tonne of CaCO₃ were US\$136 for desulfurization and US\$323 for high-purity CaCO₃ (market price of CaCO₃ of about \$400 per tCaCO₃¹⁴⁷).

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, the 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 the 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 CO₂ could be achieved.⁸² The natural weathering of old tailing piles has been studied by several groups^{193,194} (Table 5). Wilson *et al.*¹⁹³ estimated that the chrysotile in the tailing piles had carbonated approximately 0.3% per year. It was suggested that the CO₂ 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 CO₂ pressures and achieved a maximum carbonate conversion of 0.5% after 5 h at 375 °C in a 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 were only US\$600–1600 per tCO₂.¹⁶³ 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 annually¹⁹⁷).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), anatase (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* a direct process route at ambient temperatures and pressures^{162,166,167} and a sequestration capacity of 0.04–0.05 tCO₂ per t RM has been reported (Table 5). The roughly calculated cost of CO₂ sequestration is at US\$147 per 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 the pH of the slurry to a less hazardous level and capturing in the process 0.030–0.035 tCO₂ per t of RM.¹⁹²

As the primary source of alkalinity in RM is NaOH, the main carbonation products are Na₂CO₃ and NaHCO₃.¹⁶² Soluble Na-carbonates provide a less permanent CO₂ storage than solid Ca-Mg-carbonates because of their solubility. In order to provide a more permanent CO₂ 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 CO₂ 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 CO₂ 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 an E_{CO_2} of 16–27% (i.e. 0.16–0.27 tCO₂ per t-APMW) at 20–30 °C and 2–10 bar over 2–72 h.^{149,168}

4. Product 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 > 93 wt% can be produced by indirect processes.^{72,163,201} Also, 99 wt% pure calcite with particles > 5 μm 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 of carbonate products can be divided into 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 filler applications seem to be the most appropriate for silica and carbonate 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 agents to buffer the acidity of soils are promising, but require the MC products to be free from potential pollutants that might derive from particular flue gas or mineral wastes converted into 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 the amorphous phase and with particles smaller than < 30 μm, 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 MC.¹⁰

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 the form 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 the trigonal, orthorhombic, and hexagonal crystal system, respectively.²⁰⁶ 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 ink.²⁰⁸

By controlling the initial concentration of the reagents, stirring speed, pH, type and amount of additives, and other reaction conditions, CaCO₃ with different polymorphs, morphologies, and grain sizes can be obtained. For example, different CaCO₃ 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 μm in length) upon changing the operating variables, CaO concentration, the CO₂ flow rate and surfactants concentration.²⁰³ Addition of the ethyltrimethyl ammonium bromide cationic surfactant (2%) produced narrow size rounded particle morphologies either rhombohedral or spherical and a limited amount of agglomerate.²⁰³

Finally, enzymes such as Carbonic Anhydrase (CA) have been used to enhance carbonation efficiency and modify the properties of the reaction products. Mesoporous alumina synthesized from the egg shell membrane and pore-expanded SBA-15 was used as a template to immobilise CA.^{209,210} The carbonation capacity of alumina immobilized CA was found to be 25% lower compared to that obtained in the 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 by 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 an 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 by conventional methods. However, a very small liquid ratio (5 g L⁻¹) was required to get high dissolution efficiency (73%), rendering this method expensive because of the large reactor volumes required.²¹² Also, calcium carbonate powder produced from steel slag presented inferior brightness compared to traditional PCC (mainly due to iron and manganese content) 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 the goethite–calcite nanocomposite has been proposed. This synthesis involved the sequential precipitation of (1) nanosized acicular goethite (α-FeOOH); (2) the instantaneous precipitation of portlandite (Ca(OH)₂) by adding the CaCl₂ salt to a goethite alkaline suspension (2NaOH + CaCl₂ = Ca(OH)₂ + 2NaCl) and; (3) sub-micrometric calcite precipitation by injection of CO₂ into a goethite–portlandite alkaline suspension (Ca(OH)₂ + CO₂ = CaCO₃ + H₂O).²¹⁴ The precipitated nanocomposite had a surface area of around 92 m² g⁻¹ when synthesized at 30 °C and 45 m² g⁻¹, when synthesized at 70 °C.

It has to be mentioned that other methods have been proposed to convert CO₂ into chemicals and fuels. Compared to utilisation of MC products as construction or filling materials, which could in theory absorb Gt of CO₂, industrial utilization of CO₂ as solvent and reactant amounts to only 0.5 wt% (128 Mt per year) of the total anthropogenic CO₂ 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 CO₂ to hydrocarbons has been extensively reviewed.^{215–217} Even if technologies have been developed for large-scale CO₂ 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 CO₂ consumption is presented in Table 6.

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

CO ₂ uses	Existing (future) CO ₂ demand (Mt per 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)

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 CO₂ 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 criteria including the CO₂ emissions in the act of reusing it.¹⁹²

5. Summary and remarks

It is generally accepted that to reduce the level of CO₂ 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 CO₂, 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 CO₂.

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/Oman.²¹⁸ 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 a 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 1 Mt per year 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 and Ca-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 CO₂ mineralisation. Regardless of several benefits, such as avoiding costs for mining and transport, the current CO₂ mineralization technologies developed for wastes still cannot compete with geological storage in terms of potential quantity and cost of sequestered CO₂. The ones that appear to carbonate easily under mild conditions (contain free lime, do not require additional grinding, bind CO₂ 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 CO₂ emitted from OS based heat and power sector¹⁶¹).

Overall, the processes that are attracting major attention (Fig. 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 per tCO₂.²²⁰

Direct gas-solid processes, which require temperatures up to 500 °C and fine grinding of minerals (5–35 μm), achieve low capture efficiency and are not viable on the 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 shows significant improvements in 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, no public data are available to estimate the potential deployment and costs associated with these processes.

Multistep aqueous indirect processes in the 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 kinetics in the presence of additives. However, the energy intensive chemical 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 their 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 a few demonstration plants: the first is the Calera process, in the

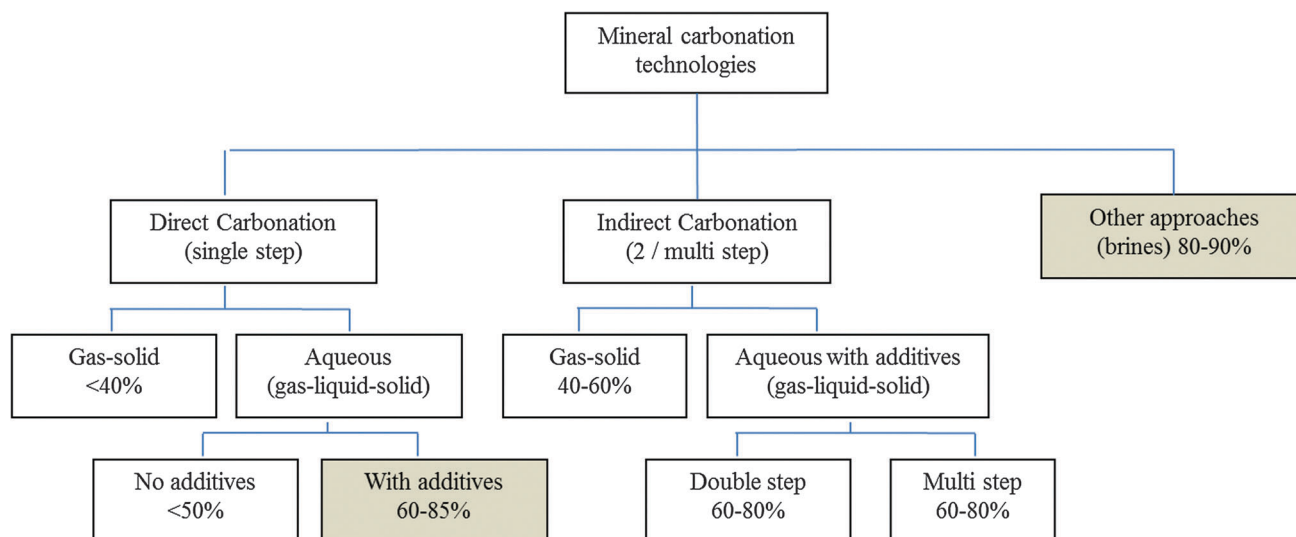


Fig. 9 Mineral carbonation process routes. In dark the most promising technologies at the current state of research and development.⁸



gas fired Moss Landing plant (USA), which has been running for about two years. The plant showed the technical capacity to capture CO₂ (30 kt per year) from a 10 MW 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 a 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 a few projects based on inorganic wastes have moved to the 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 CO₂, SO₂ and Hg emissions in Point of Rocks, USA.^{156,221} Also, accelerated carbonation has been applied for the commercial production of aggregates from APC residues²²² 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 per year generated in a nearby ammonia plant.¹⁹² However, ~30 t red mud per tCO₂ is 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 it is feasible to sequester more than 80% of 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 fuel resources.²²⁶

5.3 MC cost assessment

One of the major challenges for CCS including MC projects is the cost. Recently, an estimated transport and storage cost of ~\$17 per 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, the total cost of *in situ* MC will be in the range 72–129 per tCO₂ (considering a CO₂ capture cost of \$55–112 per tCO₂), which is by far larger than recent European carbon market CO₂ price of ~\$7 per tCO₂. However, it has to be pointed out that geological storage costs do not take into account potential long term monitoring costs due to the un-reactivity of dry CO₂ in sedimentary rocks. Also, 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 \$ per tCO₂ for olivine, wollastonite and serpentine, respectively.⁸² Recent developments in the NETL process estimated an overall mineral carbonation process cost of A\$ 70 per tCO₂ avoided, where the direct use of thermal heat instead of electrical energy, coupled to partial dehydroxylation with heat integration led to a 63% decrease in energy requirement for thermal-activation.^{91–93}

Huijgen *et al.*⁷⁰ estimated a sequestration cost (based on depreciation of investments and variable and fixed operating costs) of €102 per tCO₂ 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 per tCO₂, depending on the operation conditions (spray trickle bed systems in air²²⁷ or at 200 °C, 20 bar, 100% CO₂⁷⁰).

Using the indirect aqueous carbonation route with production of value-adding products such as high-purity PCC or hydro-magnesite from EAF slags or serpentine would require chemicals (HCl, HNO₃, CH₃COOH, NaOH) for a cost of \$600–4500 per 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 per tCO₂ 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 per t-CaCO₃, depending on product purity.¹⁴⁷

As already discussed, these costs are still higher compared to geological storage cost.²²⁸ 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, capture costs applied to chemicals, fertilizers, refineries and gas fuelled plants could be up to \$235 per tCO₂ due 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 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



Table 7 Potential uses of carbonation products from some MC processes²³⁰

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

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 a carbon neutral building block (CO₂, sand, cement) manufactured by Lignacite, 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 on a 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 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 CO₂ in terms of CO₂ avoided, energy required to obtain the carbonation products and market values. Although these “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 maintain 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 CO₂ sequestration potential of wastes remains marginal on a global scale of CO₂ emissions.

5.4 Final remarks

Despite the large resources available for CO₂ 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 development 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 the 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.

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References

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



- 2 IPCC, ed. R. K. Pachauri and A. Reisinger, IPCC, Geneva, Switzerland, 2007, p. 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. Fernández, 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 Environ. Sci.*, 2014, **7**, 130–189.
- 5 B. Praetorius and 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, ed. B. Metz, O. Davidson, H. C. de Coninck, M. Loos and L. A. Meyer, Cambridge University Press, Cambridge, United Kingdom and New York, NY, USA, 2005, p. 442.
- 10 A. Sanna, M. R. Hall and M. M. Maroto-Valer, *Energy Environ. Sci.*, 2012, **5**, 7781–7796.
- 11 G. Holmes and D. W. Keith, *Philos. Trans. R. Soc., A*, 2012, **370**, 4380–4403.
- 12 K. Z. House, A. C. Baclig, M. Ranjan, E. A. van Nierop, J. Wilcox and 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 Environ. Sci.*, 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, R. Kleerebezem, H. M. Jonkers, G.-J. Witkamp and M. C. M. van Loosdrecht, *Trends Biotechnol.*, 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 and S. Vijayakumari, *Energy Procedia*, 2011, **4**, 2885–2892.
- 20 H. Yan, J. Y. Zhang, Y. C. Zhao and C. G. Zheng, *Sci. China: Technol. Sci.*, 2013, **56**, 2219–2227.
- 21 A. Olajire, *J. Pet. Sci. Eng.*, 2013, **109**, 364–392.
- 22 R. Zevenhoven, J. Fagerlund and J. K. Songok, *Greenhouse Gases: Sci. Technol.*, 2011, **1**, 48–57.
- 23 E. R. Bobicki, Q. Liu, Z. Xu and H. Zeng, *Prog. Energy Combust. Sci.*, 2012, **38**, 302–320.
- 24 S.-Y. Pan, E. E. Chang and P.-C. Chiang, *Aerosol Air Qual. Res.*, 2012, **12**, 770–791.
- 25 A. Sanna, M. Dri, M. R. Hall and M. Maroto-Valer, *Appl. 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, <http://www.netl.doe.gov/>.
- 28 R. Berner, A. Lasaga and R. Garrels, *Am. J. Sci.*, 1983, **283**, 641–683.
- 29 S. Gislason, E. Oelkers, E. Eiriksdottir, M. Kardjilov, G. Gisladottir, B. Sigfusson, A. Snorrason, S. Elefsen, J. Hardardottir, P. Torssander and N. Pskarsson, *Earth Planet. Sci. Lett.*, 2009, **277**, 213–222.
- 30 B. M. Reinhardt, *Schweiz. Mineral. Petrogr. Mitt.*, 1969, **49**, 1–30.
- 31 F. Boudier and R. G. Coleman, *J. Geophys. Res.*, 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 and J. Havig, *Chem. Geol.*, 2012, **330/331**, 86–100.
- 34 G. D. Nicholls, *Mineral. Mag.*, 1965, **34/268**, 373–388.
- 35 C. Dessert, B. Dupre', J. Gaillardet, L. Francois and C. Allegre, *Chem. Geol.*, 2003, **202**, 257–273.
- 36 E. Rosenthal, *J. Hydrol.*, 1987, **89/3–4**, 329–352.
- 37 J. Matter and T. Takahashi, *Geochem., Geophys., Geosyst.*, 2007, **8/2**, 1–19.
- 38 K. House, D. Schrag, C. Harvey and K. Lackner, *PNAS*, 2006, **103**, 12291–12295.
- 39 P. Olsen, D. Kent, B. Cornet, W. Witte and R. Schlische, *Geol. Soc. Am. Bull.*, 1996, **108/1**, 40–77.
- 40 B. McGrail, A. Spane, E. Sullivan, D. Bacon and G. Hund, *Energy Procedia*, 2011, **4**, 5653–5660.
- 41 N. Zakharova, D. Goldberg, C. Sullivan, M. Herron and J. Grau, *Geochem., Geophys., Geosyst.*, 2012, **13/11**, 1–22.
- 42 T. Flaathen, S. Gislason, E. Oelkers and A. Sveinbjornsdottir, *Appl. Geochem.*, 2009, **24**, 463–474.
- 43 S. Gislason, D. Wolff-Boenisch, A. Stefansson, E. Oelkers, E. Gunnlaugsson, H. Sigurdardottir, B. Sigfusson, W. Broecker, J. Matter, M. Stute, G. Axelsson and T. Fridriksson, *Int. J. Greenhouse Gas Control*, 2010, **4**, 537–545.
- 44 J. Matter and P. Kelemen, *Nat. Geosci.*, 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 and 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 and 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 and S. Poag, *U.S. Geol. Surv. Circ.*, 1978, **773**, 1–28.
- 51 T. Maguire, R. Sheridan and R. Volkert, *Geodynamics*, 2004, **37**, 457–485.
- 52 D. Goldberg, D. Kent and P. Olsen, *PNAS*, 2010, **107/4**, 1327–1332.



- 53 D. Goldberg and A. Slagle, *Energy Procedia*, 2009, **1**, 3675–3682.
- 54 W. K. O'Connor, D. C. Dahlin, G. E. Rush, S. J. Gerdemann, L. R. Penner and D. N. Nilsen, Albany Research Centre, DOE/ARC-TR-04-002, USA, 2005.
- 55 O. Levenspiel, *Chemical Reaction Engineering*, John Wiley & Sons, New York, 2nd edn, 1972.
- 56 S. Teir, H. Revitzer, S. Eloneva, C.-J. Fogelholm and R. Zevenhoven, *Int. J. Miner. Process.*, 2007, **83**, 36–46.
- 57 T. A. Haug, R. A. Kleiv and I. A. Munz, *Appl. Geochem.*, 2010, **25**, 1547–1563.
- 58 M. Hänchen, S. Krevor, M. Mazzotti and K. Lackner, *Chem. Eng. Sci.*, 2007, **62**, 6412–6422.
- 59 M. Hänchen, V. Prigiobbe, G. Storti, T. M. Seward and M. Mazzotti, *Geochim. Cosmochim. 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, 18th Annual International Pittsburgh coal conference, Newcastle, Australia, 2001.
- 62 B. Bonfils, C. Julcour-Lebigue, F. Guyot, F. Bodéan, P. Chiquet and F. Bourgeois, *Int. J. Greenhouse Gas Control*, 2012, **9**, 334–346.
- 63 D. Daval, I. Martinez, J. Corvisier, N. Findling, B. Goffé and F. Guyot, *Chem. Geol.*, 2009, **262**, 262–277.
- 64 G. J. Stockmann, D. Wolff-Boenisch, S. R. Gislason and E. H. Oelkers, *Chem. Geol.*, 2013, **337–338**, 56–66.
- 65 R. M. Santos and T. Van Gerven, *Greenhouse Gases: Sci. Technol.*, 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 and 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, Proceedings of the 25th International Technical Conference on Coal Utilization & Fuel Systems, Clear Water, FL, 2000.
- 70 W. J. J. Huijgen, R. N. J. Comans and G.-J. Witkamp, *Energy Convers. Manage.*, 2007, **48**, 1923–1935.
- 71 S. Arickx, T. Van Gerven and C. Vandecasteele, *J. Hazard. Mater.*, 2006, **B137**, 235–243.
- 72 X. Wang and M. M. Maroto-Valer, *ChemSusChem*, 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 Convers. Manage.*, 1997, **38**, S259–S264.
- 75 H. F. DaCosta, M. Fan and A. T. Russell, 20100221163, 2010.
- 76 S. Kwon, M. Fan, H. F. M. DaCosta and A. G. Russell, *J. Environ. Sci.*, 2011, **23**, 1233–1239.
- 77 S. Kwon, PhD dissertation, Georgia Institute of Technology, 2011.
- 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 and D. W. Hoyt, *Int. J. Greenhouse Gas Control*, 2011, **5**, 1081–1092.
- 80 W. J. J. Huijgen, G. J. Ruijg, R. N. J. Comans and G.-J. 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, *Environ. Sci. Technol.*, 2007, **41**, 2587–2593.
- 83 O. S. Pokrovsky and J. Schott, *Geochim. Cosmochim. Acta*, 1999, **63**, 881–897.
- 84 W. J. J. Huijgen and R. N. J. Comans, ECN-C-05-022, Energy Research Centre of The Netherlands, Petten, The Netherlands, 2005.
- 85 W. K. O'Connor, D. C. Dahlin, G. E. Rush, S. J. Gerdemann, L. R. Penner and R. P. Nielsen, Albany Research Center, DOE/ARC-TR-04-002, Albany, OR, USA, 2004.
- 86 S. J. Gerdemann, D. C. Dahlin, W. K. O'Connor, 6th International Conference on Green House Gas Control Technologies, Kyoto, Japan, 2002.
- 87 M. Fabian, M. Shopska, D. Paneva, G. Kadinov, N. Kostova, E. Turianicova, J. Briancin, I. Mitov, R. A. Kliev and P. Balaz, *Miner. Eng.*, 2010, **23**, 616–620.
- 88 P. Balaz, E. Turianicova, M. Fabian, R. A. Kliev, J. Briancin and A. Obut, *Int. J. Miner. Process.*, 2008, **88**, 1–6.
- 89 Chizmeshya *et al.*, 2nd Annual Conference on Carbon Capture & Sequestration, Pittsburgh, 2002.
- 90 P. Raschman, A. Fedoročková and G. Sučík, *Hydrometallurgy*, 2013, **139**, 149–153.
- 91 H. Boerrigter, WO2008142025 A2, 2008.
- 92 A. Fedoročková, M. Hreus, P. Raschman and G. Sučík, *Miner. Eng.*, 2012, **32**, 1–4.
- 93 R. D. Balucan, B. Z. Dlugogorski, E. M. Kennedy, I. V. Belovac and G. E. Murch, *Int. J. Greenhouse Gas Control*, 2013, **17**, 225–239.
- 94 A. Sanna, X. Wang, A. Lacinska, M. Styles, T. Paulson and M. M. Maroto-Valer, *Miner. Eng.*, 2013, **49**, 135–144.
- 95 G. F. Brent, 20090305378, 2009.
- 96 GCCS, Accelerating the Uptake of CCS: Industrial Use of Captured Carbon Dioxide, P. Brinckerhoff, 2011, p. 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.*, US8062418 B2, 2011.
- 100 B. R. Constantz, *et al.*, US2009001020 A1, 2009.
- 101 J. D. Jones, *US Pat.*, 7,727,374, Skyonic Corporation, 2010.
- 102 Skyonic, 2013, <http://skyonic.com/wp-content/uploads/2010/02/Skyonic-Groundbreaking-Release-September-30-2013.pdf>.
- 103 J. I. Drever and L. L. Stillings, *Colloids Surf., A*, 1997, **120**, 167–181.
- 104 J. Declercq, O. Bosc and E. H. Oelkers, *Appl. Geochem.*, 2013, **39**, 69–77.
- 105 A. A. Olsen and J. D. Rimstidt, *Geochim. Cosmochim. Acta*, 2008, **72**, 1758–1766.
- 106 S. Eloneva, A. Said, C.-J. Fogelholm and R. Zevenhoven, *Appl. 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 and K. S. Lackner, *Energy Procedia*, 2009, **1**, 4867–4871.
- 111 S. C. M. Krevor and K. S. Lackner, *Int. J. Greenhouse Gas Control*, 2011, **5**, 1073–1080.
- 112 J. Baldyga, M. Henczka and K. Sokolnicka, *Mater. Lett.*, 2010, **64**, 702–704.
- 113 A.-H. A. Park, R. Jadhav and L.-S. Fan, *Can. J. Chem. Eng.*, 2003, **81**, 885–890.
- 114 A.-H. A. Park and L.-S. Fan, *Chem. Eng. Sci.*, 2004, **59**, 5241–5247.
- 115 E. Gal, WO/2006/022885, 2006.
- 116 A. Kothandaraman, PhD thesis, Massachusetts Institute of Technology, 2010, http://sequestration.mit.edu/pdf/Anusha_Kothandaraman_thesis_June2010.pdf.
- 117 X. Wang and M. M. Maroto-Valer, *Energy*, 2013, **51**, 431–438.
- 118 J. Fagerlund, E. Nduangu and R. Zevenhoven, *Energy Procedia*, 2011, **4**, 4993–5000.
- 119 J. Fagerlund, E. Nduangu, I. Romão and R. Zevenhoven, *Energy*, 2012, **41**, 184–191.
- 120 J. Fagerlund, J. Highfield and R. Zevenhoven, *RSC Adv.*, 2012, **2**, 10380–10393.
- 121 J. Highfield, H. Q. Lim, J. Fagerlund and R. Zevenhoven, *RSC Adv.*, 2012, **2**, 6535–6541.
- 122 E. Nduagu, T. Björklöf, J. Fagerlund, J. Wärnå, H. Geerlings and R. Zevenhoven, *Miner. Eng.*, 2012, **30**, 75–87.
- 123 R. Hunwick, 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, 27th International Technical Conference on Coal Utilization & Fuel Systems, Clearwater, FL, USA, 2002.
- 125 J.-H. Wee, *Appl. 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, *Environ. Sci. Technol.*, 2006, **40**, 2790–2796.
- 128 D. N. Huntzinger, J. S. Gierke, L. L. Sutter, S. K. Kawatrad and T. C. Eisele, *J. Hazard. Mater.*, 2009, **168**, 31–37.
- 129 O. Velts, M. Uibu, J. Kallas and R. Kuusik, *J. Hazard. Mater.*, 2011, **195**, 139–146.
- 130 G. Costa, R. Baciocchi, A. Poletti, R. Pomi, C. D. Hills and P. J. Carey, *Environ. Monit. Assess.*, 2007, **135**, 55–75.
- 131 E. E. Chang, C.-H. Chen, Y.-H. Chen, S.-Y. Pan and P.-C. Chiang, *J. Hazard. Mater.*, 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, *Appl. Therm. Eng.*, 2013, **57**, 154–163.
- 136 S. Eloneva, Doctoral Dissertation Doctoral Dissertation, Aalto University School of Science and Technology, 2010. <https://aaltodoc.aalto.fi/bitstream/handle/123456789/4876/isbn9789526034577.pdf?sequence=1>.
- 137 S. Monkman and Y. Shao, *J. Mater. Civ. Eng.*, 2006, **18**, 768–776.
- 138 F. J. Doucet, *Miner. Eng.*, 2009, **23**, 262–269.
- 139 W. J. J. Huijgen, G.-J. Witkamp and R. N. J. Comans, *Environ. Sci. Technol.*, 2005, **39**, 9676–9682.
- 140 E. E. Chang, S.-Y. Pan, Y.-H. Chen, C.-S. Tan and P.-C. Chiang, *J. Hazard. Mater.*, 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, *Int. J. 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. Poletti and R. Pomi, *Greenhouse Gases: Sci. Technol.*, 2011, **1**, 312–319.
- 145 R. M. Santos, M. Bodor, P. N. Dragomir, A. G. Vraciu, M. Vlad and T. Van Gerven, *Miner. Eng.*, 2014, **59**, 71–81.
- 146 R. M. Santos, J. Van Bouwel, E. Vandeveld, G. Mertens, J. Elsen and T. Van Gerven, *Int. J. Greenhouse Gas Control*, 2013, **17**, 32–45.
- 147 Y. Katsuyama, A. Yamasaki, A. Iizuka, M. Fujii, K. Kumagai and Y. Yanagisawa, *Environ. Prog.*, 2005, **24**, 162–170.
- 148 P. J. Gunning, C. D. Hills and P. J. Carey, *Waste Manage.*, 2009, **29**, 2722–2729.
- 149 P. J. Gunning, C. D. Hills and P. J. Carey, *Waste Manage.*, 2010, **30**, 1081–1090.
- 150 S. Teramura, N. Isu and K. Inagaki, *J. Mater. Civ. Eng.*, 2000, **12**, 288–293.
- 151 S. Kashef-Haghighi and S. Ghoshal, *Ind. Eng. Chem. Res.*, 2009, **49**, 1143–1149.
- 152 M. Fernández Bertos, X. Li, S. J. R. Simons, C. D. Hills and P. J. Carey, *Green Chem.*, 2004, **6**, 428–436.
- 153 J. Sun, M. Fernandez Bertos and S. J. R. Simons, *Energy Environ. Sci.*, 2008, **1**, 370–377.



- 154 G. Montes-Hernandez, R. Perez-Lopez, F. Renard, J. M. Nieto and L. Charlet, *J. Hazard. Mater.*, 2009, **161**, 1347–1354.
- 155 M. G. Nyambura, G. W. Mugeru, P. L. Felicia and N. P. Gathura, *J. Environ. Manage.*, 2011, **92**, 655–664.
- 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, *J. Environ. Manage.*, 2013, **127**, 212–220.
- 158 M. Back, M. Kuehn, H. Stanjek and S. Peiffer, *Environ. Sci. Technol.*, 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, *J. Environ. Manage.*, 2009, **90**, 1253–1260.
- 162 R. C. Sahu, R. K. Patel and B. C. Ray, *J. Hazard. Mater.*, 2010, **179**, 29–34.
- 163 S. Teir, S. Eloneva, C.-J. Fogelholm and R. Zevenhoven, *Appl. Energy*, 2009, **86**, 214–218.
- 164 F. Larachi, I. Daldoul and G. Beaudoin, *Geochim. Cosmochim. Acta*, 2010, **74**, 3051–3074.
- 165 S. Teir, R. Kuusik, C.-J. Fogelholm and R. Zevenhoven, *Int. J. Miner. Process.*, 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, *J. Hazard. Mater.*, 2010, **176**, 1044–1050.
- 168 R. Perez-Lopez, G. Montes-Hernandez, J. M. Nieto, F. Renard and L. Charlet, *Appl. Geochem.*, 2008, **23**, 2292–2300.
- 169 M. Uibu, O. Velts and R. Kuusik, *J. Hazard. Mater.*, 2010, **174**, 209–214.
- 170 A. A. Olajire, *Energy*, 2010, **35**, 2610–2628.
- 171 A. Iizuka, 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, *J. Hazard. Mater.*, 2004, **B112**, 193–205.
- 173 M. Uibu and R. Kuusik, *Oil Shale*, 2009, **26**, 40–58.
- 174 D. K. Lee, *Chem. Eng. J.*, 2004, **100**, 71–77.
- 175 S.-M. Shih, C. u.-S. Ho, Y.-S. Song and J.-P. Lin, *Ind. Eng. Chem. Res.*, 1999, **38**, 1316–1322.
- 176 S. N. Lekakh, C. H. Rawlins, D. G. C. Robertson, V. L. Richards and K. D. Peaslee, *Metall. Mater. 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 Convers. Manage.*, 2005, **46**, 687–699.
- 179 M. Dri, A. Sanna and M. M. Maroto-Valer, *Fuel Process. Technol.*, 2013, **113**, 114–122.
- 180 M. Dri, A. Sanna and M. M. Maroto-Valer, *Appl. Energy*, 2014, **113**, 515–523.
- 181 L. Wang, Y. Jin and Y. Nie, *J. Hazard. Mater.*, 2010, **174**, 334–343.
- 182 X. Li, M. Fernandez Bertos, C. D. Hills, P. J. Carey and S. Simon, *Waste Manage.*, 2007, **27**, 1200–1206.
- 183 E. Rendek, G. Ducom and P. Germain, *J. Hazard. Mater.*, 2006, **B128**, 73–79.
- 184 T. Van Gerven, E. Van Keer, S. Arickx, M. Jaspers, G. Wauters and C. Vandecasteele, *Waste Manage.*, 2005, **25**, 291–300.
- 185 J. A. Meima, R. D. van der Weijden, T. T. Eighmy and R. N. J. Comans, *Appl. Geochem.*, 2002, **17**, 1503–1513.
- 186 A. Polettini and R. Pomi, *J. Hazard. Mater.*, 2004, **B113**, 209–215.
- 187 R. Baciocchi, G. Costa, E. Lategano, C. M. Polettini, R. Pomi, P. Postorino and S. Rocca, *Waste Manage.*, 2010, **30**, 1310–1317.
- 188 R. Baciocchi, A. Polettini, R. Pomi, V. Prigiobbe, V. Von Zedwitz and A. Steinfeld, *Energy Fuels*, 2006, **20**, 1933–1940.
- 189 H. Ecke, *Waste Manage.*, 2003, **23**, 631–640.
- 190 Y. Soong, D. L. Fauth, B. H. Howard, R. J. Jones, D. K. Harrison, A. L. Goodman, M. L. Gray and E. A. Frommell, *Energy Convers. Manage.*, 2005, **47**, 1676–1685.
- 191 D. N. Huntzinger and T. D. Eatmon, *J. Cleaner Prod.*, 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-use-captured-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, *Econ. Geol.*, 2009, **104**, 95–112.
- 194 H. C. Oskierski, B. Z. Dlugogorski and G. Jacobsen, *Chem. Geol.*, 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, *Miner. Eng.*, 2013, **48**, 116–125.
- 196 M. Hitch, S. M. Ballantyne and S. R. Hindle, *Int. J. Min., Reclam. Environ.*, 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, *J. Hazard. Mater.*, 2010, **182**, 710–715.
- 199 C. Si, Y. Ma and C. Lin, *J. Hazard. Mater.*, 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 and R. Zevenhoven, *Energy Convers. Manage.*, 2005, **46**, 2954–2979.
- 202 A.-H. Park and L.-S. Fan, US7,722,842 B2, 2010.
- 203 S. M. El-Sheikh, S. El-Sherbiny, A. Barhoum and Y. Deng, *Colloids Surf., A*, 2013, **422**, 44–49.
- 204 P. Gronchi, T. De Marco and L. Cassar, US6,716,408 B1, 2004.
- 205 A. Sanna, in *Fossil Fuels: Sources, Environmental Concerns and Waste Management Practices*, ed. R. Kumar, Nova Science Pub. Inc., New York, 2013, ch. 6, pp. 199–208.



- 206 R. E. F. Lindeboom, I. Ferrer, J. Weijma and J. B. van Lier, *Water Res.*, 2010, **47**, 3742–3751.
- 207 C. Y. Tai and F. B. Chen, *AIChE J.*, 1998, 1790–1798.
- 208 G. Li, Z. Li and H. Ma, *Int. J. Miner. Process.*, 2013, **123**, 25–31.
- 209 S. Wanjari, C. Prabhu, N. Labhsetwar and S. Rayalu, *J. Mol. Catal. B: Enzym.*, 2013, **93**, 15–22.
- 210 N. Favre, M. L. Christ and A. C. Pierre, *J. Mol. Catal. B: Enzym.*, 2009, **60**, 163–170.
- 211 P. C. Sahoo, Y. N. Jang and S. W. Lee, *J. Cryst. Growth*, 2013, **373**, 96–101.
- 212 A. Said, H.-P. Mattila, M. Järvinen and R. Zevenhoven, *Appl. Energy*, 2012, **112**, 765–771.
- 213 M. Faatz, F. Gröhn and G. Wegner, *Adv. Mater.*, 2004, **16**, 996–1000.
- 214 G. Montes-Hernandez, F. Renard, R. Chiriac, N. Findling, J. Ghanbaja and F. Toche, *Chem. Eng. J.*, 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. Pérez-Ramírez, *Energy Environ. Sci.*, 2013, **6**, 3112–3135.
- 217 J. Qiao, Y. Liu, F. Hong and J. Zhang, *Chem. Soc. Rev.*, 2014, **43**, 631–675.
- 218 W. S. Broecker, *Elements*, 2008, **4**, 4295–4297.
- 219 F. Bodéan, 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 and G. M. Dipple, *Miner. Eng.*, 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, *Environ. Chem.*, 2006, **3**, 297–303.
- 224 B. E. H. Jones and R. J. Haynes, *Crit. Rev. Environ. Sci. Technol.*, 2011, **41**, 271–315.
- 225 D. J. Cooling, in *Paste 2007—Proceedings of the Tenth International Seminar on Paste and Thickened Tailings*, Australian Center for Geomechanics, ed. A. B. Fourie and R. J. Jewell, Perth, Australia, 2007, pp. 3–15.
- 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 Convers. Manage.*, 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. <http://www.canadiancleanpowercoalition.com/>.
- 230 CSLF, 2012, Annual Meeting Documents Book, CSLF-T-2012-10, 4 September 2012, Perth, Australia.

