Sheridan College

SOURCE: Sheridan Scholarly Output Undergraduate Research **Creative Excellence**

Faculty Publications and Scholarship

School of Chemical and Environmental Sciences

9-2012

Stabilization of Basic Oxygen Furnace Slag by Hotstage Carbonation Treatment.

Rafael M. Santos

Katholieke Universiteit Leuven, rafael.santos@sheridancollege.ca

Da Ling

Katholieke Universiteit Leuven

Amin Sarvaramini

Laval University

Muxing Guo

Katholieke Universiteit Leuven

Jan Elsen

Katholieke Universiteit Leuven

See next page for additional authors

Follow this and additional works at: http://source.sheridancollege.ca/fast_chem_publ



Part of the Chemical Engineering Commons

SOURCE Citation

Santos, Rafael M.; Ling, Da; Sarvaramini, Amin; Guo, Muxing; Elsen, Jan; Larachi, Faïçal; Beaudoin, Georges; Blanpain, Bart; and Van Gerven, Tom, "Stabilization of Basic Oxygen Furnace Slag by Hot-stage Carbonation Treatment." (2012). Faculty Publications and Scholarship. Paper 20.

http://source.sheridancollege.ca/fast_chem_publ/20



This work is licensed under a Creative Commons Attribution-Noncommercial-No Derivative Works 4.0 License.

This Article is brought to you for free and open access by the School of Chemical and Environmental Sciences at SOURCE: Sheridan Scholarly Output Undergraduate Research Creative Excellence. It has been accepted for inclusion in Faculty Publications and Scholarship by an authorized administrator of SOURCE: Sheridan Scholarly Output Undergraduate Research Creative Excellence. For more information, please contact source@sheridancollege.ca.

Authors Rafael M. Santos, Da Ling, Amin Sarvaramini, Muxing Guo, Jan Elsen, Faïçal Larachi, Georges Beaudoin, Bart Blanpain, and Tom Van Gerven

Accepted Manuscript

Stabilization of basic oxygen furnace slag by hot-stage carbonation treatment

Rafael M. Santos, Da Ling, Amin Sarvaramini, Muxing Guo, Jan Elsen, Faï çal Larachi, Georges Beaudoin, Bart Blanpain, Tom Van Gerven

PII: S1385-8947(12)00895-9

DOI: http://dx.doi.org/10.1016/j.cej.2012.06.155

Reference: CEJ 9519

To appear in: Chemical Engineering Journal

Received Date: 28 March 2012 Revised Date: 19 June 2012 Accepted Date: 20 June 2012



Please cite this article as: R.M. Santos, D. Ling, A. Sarvaramini, M. Guo, J. Elsen, F. Larachi, G. Beaudoin, B. Blanpain, T. Van Gerven, Stabilization of basic oxygen furnace slag by hot-stage carbonation treatment, *Chemical Engineering Journal* (2012), doi: http://dx.doi.org/10.1016/j.cej.2012.06.155

This is a PDF file of an unedited manuscript that has been accepted for publication. As a service to our customers we are providing this early version of the manuscript. The manuscript will undergo copyediting, typesetting, and review of the resulting proof before it is published in its final form. Please note that during the production process errors may be discovered which could affect the content, and all legal disclaimers that apply to the journal pertain.

Stabilization of basic oxygen furnace slag by hot-stage carbonation treatment

Rafael M. Santos ^a, Da Ling ^b, Amin Sarvaramini ^c, Muxing Guo ^b, Jan Elsen ^d, Faïçal Larachi ^c, Georges Beaudoin ^e, Bart Blanpain ^b, Tom Van Gerven ^a,*

^a KU Leuven, Dept. Chemical Engineering, Willem de Croylaan 46, 3001 Leuven, Belgium

^b KU Leuven, Dept. Metallurgy and Materials Engineering, Kasteelpark Arenberg 44, 3001 Leuven, Belgium

^c Laval University, Dept. Chemical Engineering, 1065 Avenue de la Médecine, Québec, QC, Canada G1V 0A6

^d KU Leuven, Dept. Earth and Environmental Sciences, Celestijnenlaan 200e, 3001 Leuven, Belgium

^e Laval University, Dept. Geology and Geological Engineering, 1065 Avenue de la Médecine, Québec, QC,

Canada G1V 0A6

*Corresponding author. Tel:. +32 16 322342; fax: +32 16 322991.

Email addresses: rafael.santos@cit.kuleuven.be (R.M. Santos); faical.larachi@gch.ulaval.ca (F. Larachi); thomas.vangerven@cit.kuleuven.be (T. Van Gerven).

ABSTRACT

Treatment and disposal of Basic Oxygen Furnace (BOF) slag, a residue of the steel production process characterized by high basicity and propensity for heavy metal leaching, is a costly burden on metallurgical plants; a sustainable valorization route is desired. The stabilization of BOF slag utilizing hot-stage carbonation treatment was investigated; this approach envisions carbonation during the hot-to-cold pathway followed by the material after the molten slag is poured and solidified. Three experimental

methodologies were employed: (i) in-situ thermogravimetric analyzer (TGA) carbonation was used to assess carbonation reaction kinetics and thermodynamic equilibrium at high temperatures; (ii) pressurized basket reaction carbonation was used to assess the effects of pressurization, steam addition and slag particle size; and (iii) atmospheric furnace carbonation was used to assess the effect of carbonation on the mineralogy, basicity and heavy metal leaching properties of the slag. Free lime was found to be the primary mineral participating in direct carbonation of BOF slag. Initial carbonation kinetics were comparable at temperatures ranging from 500 to 800 °C, but higher temperatures aided in solid state diffusion of CO₂ into the unreacted particle core, thus increasing overall CO₂ uptake. The optimum carbonation temperature of both BOF slag and pure lime lies just below the transition temperature between carbonation stability and carbonate decomposition: 830-850 °C and 750-770 °C at 1 atm and 0.2 atm CO₂ partial pressures, respectively. Pressurization and steam addition contribute marginally to CO₂ uptake. CO₂ uptake progressively decreases with increasing particle size, but basicity reduction is similar independent of particle size. The solubility of some heavy metals reduced after carbonation (barium, cobalt and nickel), but vanadium and chromium leaching increased.

Keywords: Mineral carbonation; Steel slag; Basic Oxygen Furnace; Hot-stage; In-situ; Heavy metal leaching

1. Introduction

Integrated carbon steel production consists of ironmaking in the Blast Furnace (BF), steelmaking in the Basic Oxygen Furnace (BOF), and continuous casting of steel billets, slabs and blooms. For over a century, with iron and steel industry booming worldwide, a vast amount of slag has been produced as an inevitable by-product of the steelmaking process. While valuable applications have been found for BF slag, mostly in the construction sector such as in cement manufacturing and as a cement replacement in concrete, much of BOF slag production, estimated at 60-120 kg/t steel presently, still ends up in landfill sites [1-3]. The traditional use of BOF slag in road construction, as an aggregate, base or sub-base coarse,

has been restricted due to the slag's undesirable expansive nature, resulting in rapid deterioration of the roads [4]. The volume expansion (up to 10% [5]) is attributed to the short term hydration and the long term carbonation of free lime (CaO) and magnesium oxide (MgO) content [6].

To date, four forms of treatment have been commercially used to stabilize BOF slag: (i) weathering in slag pits to convert free lime into hydrated lime, which has been found slow and inefficient [7]; (ii) steam hydration of the slag, utilized by Nippon Steel [8]; (iii) additions of SiO₂ and O₂ (to keep the slag molten) to the slag pot, which act to dissolve the free lime and chemically bond it as silicate, utilized by Thyssen Krupp [7]; and (iv) control of the slag cooling path to stabilize the C₃S (tri-calcium silicate) phase (which otherwise is transformed into C₂S, di-calcium silicate, plus free lime), utilized by Baosteel [9,10].

An alternative and possibly more attractive and low-cost route is the reaction of the alkaline oxides with CO₂, leading to the formation of geochemically stable carbonates (e.g. CaCO₃) [11]. Besides capture of CO₂, desirable for emissions reduction, mineral carbonation has also been reported to yield positive effects in terms of the leaching behavior of alkaline earth metals, heavy metals and metalloids from steel slag [12], which can lead to further valorization of the waste material. Numerous studies in recent years have assessed the potential of steel slag carbonation for storage of CO₂ utilizing a variety of aqueous carbonation routes, including slurry carbonation [13,14], wet carbonation [15], block carbonation [16], and two-step carbonation (i.e. leaching of Ca and Mg followed by carbonation of the extracted solution) [17]. However, the major drawbacks of these approaches include the required energy intensive steps (e.g. crushing/milling of the iron-rich monolith slag, prolonged mixing, pressurization, separation, and regeneration of extractants) and the generation of products having low market demand (e.g. carbonated powdery materials) or that are still troublesome (e.g. destabilized heavy-metal containing residual solids, and salt and heavy metal laden wastewater).

To overcome the aforementioned disadvantages of conventional mineral carbonation a novel approach is herein explored, hot-stage carbonation. This approach utilizes two domains of process intensification, namely functional (thus maximizing synergistic effects from partial processes) and temporal (thus optimizing the driving forces and maximizing the specific area to which these apply) [18]. The principal concept is to take advantage of the high temperature source of the slag, and perform mineral

carbonation during slag cooling at still high (optimized) temperatures, where the kinetics are more favorable and the (thermal) energy is freely available. In particular, it is envisaged that in order to carbonate the slag while (a) it is still hot, (b) it is solidified, and (c) the material surface area is compatible with requirement for carbonation (i.e. not a monolith), a practical approach to hot-stage carbonation would be to apply it concurrently with hot-stage granulation (see Fig. SC-1 in the Supplementary Content for a conceptual scheme).

High temperature carbonation is already applied in CO₂ capture and separation systems that utilize lime-based sorbents subject to looping carbonation/calcination cycles, which typically conduct carbonation in the flue gas temperature range of 600 to 700 °C at atmospheric pressure [19-21]. Prigiobbe et al. [22] tested high temperature carbonation of air pollution control residues in the temperature range of 350-500 °C, obtaining fast carbonation kinetics (50% conversion in less than one minute) and high conversions (nearly 80%). Mikhail and Turcotte [6] were the first investigators to report carbonation of BOF slag at high temperature; while limited, their findings provided proof-of-concept for this approach. BOF slag was carbonated at 550 °C in moist CO₂ in a thermogravimetric analyzer (TGA), and 8% mass gain was attributed to CaCO₃ formation. More recently, Kao [23] studied the high temperature carbonation of BOF slag for the purpose of using the slag as a CO₂ sorbent. Breakthrough curves of the CO₂ content of the outlet gas from a tubular reactor, analyzed by infra-red (IR) detection, were reported, with optimum conditions identified as being 500 °C, 40% CO₂ inlet concentration and 10% inlet gas relative humidity. Characteristics of the carbonated solids were not extensively assessed. Yu and Wang [24] found BOF slag to carbonate well (up to 20% calcium utilization after 20 minutes reaction time) at temperatures between 500 and 550 °C, using CO₂ concentrations between 10-100%; however data and analyses were insufficient to provide clear insight on mineralogical effects or carbonation mechanisms and kinetics.

The present work presents a detailed and systematic study of high temperature BOF slag carbonation, with particular emphasis on carbonation kinetics and conversion. The effect of several process parameters (particle size, temperature, pressure, time and steam addition) on CO₂ uptake and free lime conversion are tested. In addition, the chemical behavior of the treated materials, namely basicity and heavy metal leaching, is assessed. The investigation is centered on the concept of hot-stage carbonation of the slag, that

is, experimental methods are designed to simulate the hot-to-cold stage path of the just-produced slag and to elucidate the susceptibility of the slag to carbonation along this route. To achieve these objectives, three hot-carbonation methodologies are applied, which provide varying levels of insight into the technology.

2. Materials and Methods

2.1. BOF Slag

2.1.1 Slag Feedstock

Two freshly produced BOF slag batches were used, containing differing amounts of free lime; they will be referred to as BOF_{II} (lower free lime) and BOF_{II} (higher free lime). The slags, having been solidified under production conditions, were crushed (jaw crusher) and sieved to four particle size fractions for carbonation studies: < 0.08 mm, 0.08-0.5 mm, 0.5-1.0 mm and 1.0-1.6 mm. These particle sizes were chosen as they are analogous to values reportedly produced during BOF slag dry granulation ([10,25]). The three coarser fractions were wet-sieved to remove fines; calcium saturated water was used to limit dissolution. Finely milled samples were also produced using a McCrone Micronizing Mill. Pure oxide minerals (CaO, 99% purity; Fe₂O₃, >99% purity; MgO, 98% purity; Sigma Aldrich) were also utilized for carbonation for comparison purposes.

2.1.2 Slag Characterization Methodology

Particle size analysis was performed by laser diffraction (LD, Malvern Mastersizer). Specific surface area (S_{BET}) was measured by nitrogen adsorption (Micromeritics TRISTAR 3000) using the BET (Brunauer–Emmett–Teller) model. Material specific density was measured using a pycnometer (Micromeritics AccuPyc 1330). The chemical composition of each BOF slag batch was determined by X-ray Fluorescence (XRF, Panalytical PW2400), and the mineral composition was determined by X-ray Diffraction (XRD, Philips PW1830) with Rietveld refinement technique (Topas Academic v4.1). The distribution of the mineral phases in BOF slag was observed by analysis of the polished cross-sections of

millimetre-sized particles of BOF_I slag by Scanning Electron Microscopy (SEM) coupled with Backscattered Electrons (BSE) and Energy Dispersive X-ray Spectroscopy (EDX) (Philips XL30 FEG).

2.2. High Temperature Carbonation Methodology

To study the carbonation reaction of BOF slag at high temperature, three experimental approaches were used to provide different insights into the effects of material properties and process conditions on the CO₂ uptake, carbonation conversion of the mineral phases, basicity and heavy metal leaching stabilization. The thermodynamics of the relevant carbonation reactions were assessed by FactSage 6.2 modeling.

2.2.1 In-situ TGA Carbonation

The first experimental methodology made use of a thermogravimetric analyzer (Perkin Elmer Lab System Diamond TGA-DTA) to conduct in-situ studies of BOF slag carbonation; this method and the equipment used were adapted from the methodology applied in Larachi et al. [26] for the carbonation of chrysotile mining residues. Three variations of this experimental method were used (depicted in Fig. SC-2). All experiments used approximately 50 mg of sample (finely-milled BOF slags, or reference pure materials CaO, MgO and Fe₂O₃), which was placed in a tared alumina crucible. All experimental variations commenced with a calcination step to produce 'fresh slag' as is produced industrially from the hot-stage process. This step consisted in heating the sample to 900 °C in nitrogen atmosphere (100 ml/min gas flow rate) at 20 °C/min heating rate to decompose pre-existing hydrates/hydroxides/carbonates from the material. It should be noted that as all samples underwent this initial calcination step, they experienced equal thermal histories (an important distinction as reported by Bhatia and Perlmutter [27]). Following calcination, experiments denominated as "hot-to-cold TGA carbonation" consisted in cooling the samples from 900 °C to 200 °C in 20% or 100% carbon dioxide atmosphere (100 ml/min gas flow rate) at 5 °C/min cooling rate to carbonate the material. Several flue gas streams in steel plants contain roughly 20% CO₂ [28-29], therefore this value was used for comparison with pure CO₂, which could also be produced from flue gas using one of several established CO₂ separation processes. Experiments denominated as "cold-to-

hot TGA carbonation" consisted in cooling the samples from 900 °C to 80 °C in inert nitrogen atmosphere at 20 °C/min cooling rate, followed by heating from 80 °C to 1000 °C in carbon dioxide atmosphere at 5 °C/min heating rate to carbonate the material. Finally, experiments denominated as "fixed temperature TGA carbonation" consisted in cooling the samples from 900 °C to a desired temperature (between 500 °C and 800 °C) in inert nitrogen atmosphere at 20 °C/min cooling rate, followed by holding at the desired temperature in carbon dioxide atmosphere for 60 minutes to carbonate the material. For all experiments the sample weight (loss during calcination step and gain during carbonation step) was continuously monitored to determine CO₂ uptake. A mass spectrometer with quadrupole analyzer (Thermostar Prisma QMS200, Pfeiffer Vacuum) was used to monitor the outlet gas composition and to confirm that the gas composition transition (from N₂ to CO₂) occurred quickly enough; for all practical purposes it was found that the gas composition reached nearly complete transition (> 90% CO₂) in less than one minute.

2.2.2 Pressurized Basket Reactor Carbonation

The second experimental methodology (depicted in Fig. SC-3) made use of a pressurized basket reactor to study carbonation at high pressure conditions, as well as the effect of steam addition and slag particle size. This method and the equipment used were adapted from the methodology applied in Larachi et al. [30] for the carbonation of chrysotile mining residues. The equipment is shown in Fig. SC-4 in the Supplementary Content. Approximately two grams of material was placed in a small stainless-steel mesh basket that allowed the process gas to creep into the sample material bed. The basket was slid coaxially and positioned midway in a steel cylindrical enclosure (1.65 cm internal diameter, 30 cm long) using a hollow metallic stem hosting a thermocouple driven in the middle of the bed. The cylindrical reactor was fit in an electrically-heated temperature-controlled ceramic furnace that could be operated up to 900 °C. Prior to carbonation, samples were first heated to 900 °C for calcination in argon atmosphere. Upon purging the reactor with argon flow and cooling to the desired reaction temperature (350-650 °C), carbon dioxide and steam were introduced up to the desired total pressure (up to 20 bar), consisting of 4 to 20 bar CO₂ and 0 to 8 bar steam partial pressures. Carbonation reaction was conducted for 30 minutes. Following carbonation, the reactor was immediately purged with nitrogen gas flow (300 ml/min) to preclude further carbonation.

Once purged, the reactor was heated up to 900 °C for calcination. During calcination the reactor outlet gas was directed to an infra-red (IR) detector (ABB AO2000 Uras14) for in-situ quantification of the released CO₂ amounts (i.e. integration of CO₂ gas content over time) and determination of the material CO₂ uptake. Several tests were performed in duplicate to ascertain the accuracy of the measured effects.

2.2.3 Atmospheric Furnace Carbonation

In order to produce larger amounts of high-temperature carbonated materials for XRD, basicity and leaching testing, a horizontal furnace was used. The carbonation methodology (depicted in Fig. SC-5) was analogous to the basket reactor methodology, except that up to 10 grams of material could be carbonated at a time in the furnace (it was possible to carbonate 3 grams of material in each of three alumina crucibles that could be placed simultaneously in the furnace to test different materials or particle sizes at identical process conditions, or 10 grams of a single material in a larger crucible). Carbonation experiments were conducted with 100% CO₂ gas flow at near atmospheric pressure (slight positive pressure to prevent infiltration) for varying reaction times (10 minutes to 6 hours) at fixed temperatures (200-800 °C). Following carbonation, samples were immediately withdrawn from the furnace to cease carbonation, and allowed to cool in air. The CO₂ uptake was quantified by ex-situ TGA; an amount of 30-50 mg carbonated sample was weighed in a sample pan heated from 25 to 900 °C under a nitrogen atmosphere at a heating rate of 15 °C/min. The weight loss was recorded by the microbalance and the amount of CO₂ uptake was quantified by the weight loss between 500-800 °C, which can be related to calcium carbonate decomposition [13]. Validation of the accuracy of CO₂ uptake determination via TGA and via calcination coupled to IR detection was performed by comparison of analysis results of identical samples examined with both methodologies; the average discrepancy between the two methods was 4% of the measured values (see Figs. SC-6 and SC-7 in the Supplementary Content).

2.3 Basicity and Leaching Tests

Batch leaching tests were performed on both carbonated and fresh slag samples to determine the effect of carbonation on the solubility of heavy metals and on the solution pH value. Determination of aqueous elemental concentration was performed in triplicate by Inductively Coupled Plasma Mass Spectroscopy (ICP-MS, Thermo Electron X Series). An amount of two grams of solids was mixed with 100 ml MilliQ water (L/S = 50) in a sealed polyethylene bottle, and shaken on a vibration table for 24 hours. Solution pH was measured and the solution was filtered using 0.45 µm membrane filter prior to dilution for ICP-MS measurement. Solution matrix for ICP-MS measurement was 2% nitric acid. The following elements (in atomic weight order) were analyzed: V, Cr, Co, Ni, Cu, Zn, As, Se, Mo, Cd, Sn, Sb, Ba, and Pb. Table 1 presents leaching limits (in milligrams metal leached per kilogram dry solid) of the most pertinent heavy metals for the present work (i.e. those with notable measurable concentrations and that are affected by carbonation, as reported in section 3). It was verified using fresh slags that the liquidto-solid ratio (L/S) of 50 (utilized for testing of carbonated slags due to limitation of material available) provides leaching values that are similar to, and in most cases exceeding, tests using lower L/S values (10 and 20; see Fig. SC-8 in the Supplementary Content); this is due to the solubility controlled (rather than availability controlled) equilibrium established. Therefore it is possible to compare results of tests using L/S = 50 to the leaching limits established for tests using L/S = 10 for assessment of material compliance.

3. Results and Discussion

3.1 Characterization of BOF slags

Average particle sizes by volume and specific surface areas of the two finest sieved fractions and the micronized samples are presented in Table 2. Specific densities of the slags were found to be 3.69 and 3.78 g/cm^3 for BOF_I and BOF_{II}, respectively.

Table 3 presents the elemental composition of the slags determined by XRF. It can be inferred from the calcium content that the maximum theoretical CO₂ content (in the form of calcium carbonate) of a fully

carbonated specimen would be 28.0 wt% CO_2 (equivalent to 0.39 g, CO_2 /g,slag) for BOF_1 and 31.2 wt% CO_2 (or 0.45 g, CO_2 /g,slag) for BOF_{II} .

Table 4 shows the interpreted mineral composition determined by XRD. Samples were heated to 900° C in an argon atmosphere prior to analysis to decompose pre-existing hydroxides and carbonates formed due to exposure of the samples to ambient air (i.e. natural aging). The main mineral phases of the BOF slags were identified as srebrodolskite (C_2F), the beta polymorph of di-calcium silicate (β - C_2S), and free lime (CaO). The theoretical CO_2 uptake capacity of these materials based solely on their free lime content would be 7.7 wt% CO_2 (equivalent to 0.083 g, CO_2/g ,slag) for BOF₁ and 13.9 wt% CO_2 (or 0.162 g, CO_2/g ,slag) for BOF₁.

Analysis of the polished cross-section of a millimetre-sized particle of BOF₁ slag by SEM-BSE-EDX is shown in Fig. 1. The morphology and distribution of the major mineral phases quantified by XRD can be visualized in the image. Two types of free lime are observed: precipitated free lime and residual free lime. Precipitated free lime originates from the decomposition of tri-calcium silicate to di-calcium silicate during slag cooling [33]. Residual free lime is lime that was added during the steelmaking process as a flux, and either was added in excess or not allowed enough time to dissolve [34], as a means to increase steel productivity. The type of free lime is suspected to influence the slag reactivity and propensity for instability through volume expansion [35]. Wachsmuth et al. [34] pointed out that for BOF slag containing high amounts of free lime, residual free lime is the dominant form; this was confirmed for the present samples by inspection of several slag particles by SEM-BSE-EDX. In terms of the other minor elements detected by XRF analysis (Table 3), Mn and Mg were detected in the BOF slag in solid solution with iron oxides, while Al, Ti, V and Cr were found at the grain boundaries of C₂S and residual free lime phases. These interpretations are in accordance with reported observations of Waligora et al. [35] and van Zomeren et al. [15].

3.2 High Temperature Carbonation

The desire to attempt BOF slag carbonation at high temperature stems from the fact that the carbonation thermodynamics of the alkaline components of the slag are favorable (negative Gibbs free energy of reaction change) up to high temperatures (Table 5). Lime (CaO) carbonation (Eq. 1) is theoretically possible up to 861 °C at 1 atm CO₂ pressure, while other common slag minerals such as C₂S (Eq. 2), C₂F (Eq. 3), and periclase (Eq. 4) reach carbonation/calcination equilibrium at lower temperatures of 400, 664 and 400 °C, respectively. Beyond these temperatures the reactions reverse and CO₂ is released into the gas phase, a process commonly called calcination. Carbonation susceptibility is also affected by the partial pressure of carbon dioxide. In the case of lime, reducing the CO₂ partial pressure to levels present in the atmosphere (0.0003 atm), reduces the equilibrium temperature to 511 °C. However, at levels of CO₂ typically found in flue gases of thermal processes (e.g. 0.1 atm [22]) the equilibrium temperature is significantly higher at 732 °C. Thermodynamics also suggest that pressurization can be useful, as the equilibrium temperature at 5 atm CO₂ partial pressure rises to 971 °C (this condition could be achieved, for example, using a purified CO₂ stream from flue gas via capture and concentration processes).

$$CaO + CO_2 = CaCO_3 \tag{1}$$

$$Ca2SiO4 + 2CO2 = 2CaCO3 + SiO2$$
 (2)

$$Ca_2Fe_2O_5 + 2CO_2 = 2CaCO_3 + Fe_2O_3$$
 (3)

$$MgO + CO_2 = MgCO_3 \tag{4}$$

However, favorable thermodynamics do not guarantee that these carbonation reactions will occur fast enough, as reaction kinetics also play an important role. Therefore it was the objective of this work to confirm the susceptibility of BOF slag carbonation at high temperature, both with regards to carbonation kinetics and CO₂ uptake, using three carbonation methodologies as will be detailed next.

3.2.1 In-situ TGA Carbonation

To gain insight on the interaction of BOF slag with CO₂ at high temperatures comparable to the conditions in the cooling path of the slag during metallurgical processing, in-situ TGA carbonation tests

were performed. Fig. 2 presents experimental results applying the in-situ TGA carbonation methodologies for both hot-to-cold (subfigures a and b) and cold-to-hot (subfigure c) carbonation directions. For comparison, pure CaO, MgO and Fe₂O₃ as well as micronized BOF_I and BOF_{II} slag samples were tested. The hot-to-cold carbonation direction is meant to simulate the carbonation path BOF slag would undertake in the industrial setting when being reacted immediately after slag solidification from the molten stage. Carbonation was conducted from 900 to 400 °C, with 5 °C/min cooling rate. The cold-to-hot direction, though not representative of the envisioned hot-stage carbonation technique, was also utilized for comparison and to aid gaining insight on carbonation kinetics and mechanisms. In this case the heating rate was 5 °C/min, and carbonation took place from 80 to 1000 °C. Two CO₂ partial pressures were applied: 20 vol% (balanced with N₂) and 100 vol% CO₂ at atmospheric total pressure.

It is immediately possible to conclude from Fig. 2a that carbonation extent (sample mass gain due to CO₂ incorporation as carbonates) is directly proportional to the free lime content: pure CaO achieved significantly higher uptake than the slags, of which BOF_{II}, having greater free lime content (Table 4), and slightly higher specific surface area (Table 2), carbonated more than BOF_I. In contrast the other two oxides tested (MgO and Fe₂O₃) did not carbonate in the temperature range tested (Fig. 2b). Stainless steel slag samples containing mainly calcium silicates (including C₂S) and no free lime also did not exhibit reactivity to direct carbonation at these temperatures (see Fig. SC-9 in the Supplementary Content); in contrast these materials react well in aqueous carbonation [36]. Given the large content of calcium in BOF slags (Table 3), a major part being present as C₂F (Table 4), and the comparatively low CO₂ uptakes compared to pure CaO, it can be concluded that C₂F reactivity to direct carbonation is also low, despite favorable thermodynamics (Table 5). Available literature lacks extensive data on C₂F carbonation, but Schmidt [37] found it possible to carbonate a similar mineral, Sr₂Fe₂O₅, at 800 °C, though kinetics were slow (order of hours). C₂F likely suffers from too slow kinetics to influence CO₂ uptake in the time frame of the present experiments. Therefore it is concluded that free lime is the principal mineral in BOF slag responsible for CO₂ uptake during direct high temperature carbonation. Under this premise, the carbonation conversions of CaO to CaCO₃ for BOF₁ and BOF₁₁ are 32.4% and 45.7%, respectively. These conversions on the basis of the total Ca content of the slags are equivalent to 7.0% and 16.3%, respectively.

Furthermore, while for CaO there are two distinct carbonation regimes (one kinetically controlled and a second diffusion-controlled), for the slag samples a single regime dominates. From the CaO trends, it is observable that upon commencement of carbonation, the uptake rate is sharp and linear. After this linear region, carbonation of CaO continues at a reduced rate, which levels off over time (temperature decrease) until carbonation ceases. These trends are similar at both partial pressures of CO₂, except that at lower partial pressure CaO carbonation reaches lower final uptake level. In the case of slag samples, carbonation from the start behaves similar to the second stage of CaO carbonation, that is, it is gradual and declining in rate over time. These effects can be explained by the necessity of CO₂ to diffuse across a native inert layer into slag particles from the start of carbonation to achieve further uptake, and a second regime does not occur since emerging carbonates form a dispersed patchwork within the particle, rather than a dense shell. In the case of pure CaO, significant carbonation at the exposed outer surface occurs immediately, only later to be followed by diffusion through the reacted carbonate layer into the particle core.

It is also inferred that carbonation commences at approximately the same temperature for the three reacting materials at each partial pressure: 830-850 °C at 100% CO₂, and 750-770 °C at 20% CO₂. These carbonation temperatures as a function of CO₂ partial pressure correspond to predicted values from reaction thermodynamics presented earlier (Table 5). Distinctly from CaO, however, the slag samples reached the same final carbonation extent values at both CO₂ partial pressures. This signifies that CO₂ diffusion is hindered equally by the dense unreactive mineral phases of the slags regardless of diffusion driving force (partial pressure and temperature). In the case of CaO, higher driving forces (temperature and partial pressure) and higher CO₂ concentration aid further carbonation extent, possibly by better displacement of inert gas contained in the particle pores. The indifference of slag carbonation to CO₂ partial pressure is positive for industrial implementation as it means costly and energy intensive enrichment of flue gases may not be required. Further discussion on the benefit of CO₂ pressurization is given in section 3.2.2.

Fig. 2c demonstrates that carbonation in the cold-to-hot direction differs from the hot-to-cold trends. Most striking are the higher carbonation extents achieved with all three materials (CaO and micronized BOF_I and BOF_{II}). Also, CaO begins carbonating at a lower temperature (approx. 200 °C) than the slags

(> 400 °C). Maximum carbonation extent is reached slightly prior to reaching 900 °C for the three samples. After this point, in spite of the presence of CO₂, the carbonation reaction reverses, and CaCO₃ decomposition proceeds until the sample masses return to their initial values upon holding them at 1000 °C for several minutes. The higher carbonation extents may be explained by the ever increasing CO₂ solid-state diffusivity over time, opposite to the case of hot-to-cold carbonation. In addition, it can be expected that in the hot-to-cold case, due to the higher carbonation rates at higher temperatures (more data and discussion to follow), the carbonate layer formed at the outer particle surface quickly becomes denser and less permeable than when it is formed at lower temperatures, thus deterring carbonation continuance already at an early stage. Nevertheless, as mentioned earlier, this type of carbonation methodology has a major disadvantage, as it does not take advantage of the slag cooling path but rather would require external (thermal) energy input, therefore these results are meant to aid discussion only and are not deemed industrially feasible.

As the previously performed tests applied carbonation at varying temperatures, it was not possible to infer precisely the kinetics of the reaction. Therefore fixed temperature in-situ TGA carbonation experiments were performed with both micronized slag samples (BOF_I and BOF_{II}) and pure CaO as a reference material. Results are presented in Fig. 3 for five temperatures: 500, 575, 650, 725 and 800 °C. For all experimental runs, CO₂ was introduced at the two minute mark in these plots, hence all carbonation reactions can be deemed to have started at the same time. As can be observed, carbonation of all materials at all tested temperatures commences promptly after CO₂ introduction. It can also be noted from the vertical axes that BOF_I carbonation once again achieves lower extents compared to BOF_{II}, which itself achieves lower extents than CaO at comparable temperatures. For all materials, carbonation extent is greatest at the highest temperature, 800 °C.

Similar to trends observed in the hot-to-cold carbonation experiments, CaO demonstrates dual regimes of carbonation, a rapid linear stage followed by another of diminishing reactivity. The CaO carbonation rate during the first stage is independent of temperature, and is measured to be on average 58 ± 5 mg,CO₂/g,CaO/min. In contrast, BOF slags have a very short initial steep uptake as a result of surface/outer pore carbonation (< 1 min), thereafter transitioning into the decaying-rate-type trend during

which CO_2 slowly penetrates deeper into the particle for the reaction, and thus mass gain, to continue. Because of this, temperature-controlled CO_2 diffusivity is a greater limiting factor for slag carbonation than reaction kinetics. This evidenced by the greater effect of temperature on BOF slag carbonation extent than for CaO. In the case of the latter, after 45 minutes the carbonation extent at 800 °C is 60% higher (expressed as percent mass gain) than at 500 °C. In comparison, for the slags, BOF_1 achieves nearly five times greater mass gain and BOF_{II} more than four times greater mass gain at the highest temperature over the lowest temperature.

In comparison with literature reported carbonation values for CaO in direct high temperature carbonation, the carbonation conversions achieved in the experiments so far presented in this study were lower than some studies (for example, at 800 °C, CaO carbonation conversion was 27% of the theoretical maximum carbonation extent, compared to 60-80% reported values [27,38]). Other studies, however, have also found conversion in the 20-30% range [19,39,40]. For the other two experimental methods reported herein, pressurized basket reactor carbonation and atmospheric furnace carbonation, it was verified that CaO carbonation achieved conversions over 60% at optimized process conditions. The cause of these discrepancies was identified for the present experimental system as being due to the CO₂ flow rate utilized (100 ml/min). When the flow rate was increased to 300 ml/min, the carbonation conversion at 800 °C increased to 64% (Fig. SC-10a in the Supplementary Content), though the initial carbonation rates, within the first 2.5 minutes of CO₂ introduction, were identical. This suggests that in both cases, sufficient CO₂ initially contacts the free lime surface, which carbonates equally rapidly. As the reaction progress, nitrogen gas contained within the sample crucible and inside the particle pores must be displaced by CO2 gas for the reaction to continue. However, when using a lower gas flow rate, sweeping of the displaced nitrogen is reduced, thereby CO₂ diffusion into the particles is slowed. Slower CO₂ diffusion leads to earlier occurrence of pore blockage due to carbonate formation close to the particle surface, and a limited final conversion compared to the high flow case. In the case of BOF slag, the effect of CO₂ flow rate was smaller and opposite (Fig. SC-10b in the Supplementary Content); 100 ml/min resulted in slightly higher carbonation extent. The micronized slag's relatively high surface area and presence of macropores compared to commercial CaO [40], and its dispersed free lime concentration likely attenuate the effect of

nitrogen displacement, and pore narrowing/blockage is less dependent on CO₂ flow rate. Seeing the high variability of results in the literature, it is advisable future researchers consider the effect of flow rate on their in-situ TGA carbonation experiments.

3.2.2 Pressurized Basket Reactor Carbonation

Pressurized basket reactor carbonation experiments were conducted to investigate process effects that could not be assessed with the previously presented TGA methodology. These include the effect of CO₂ pressurization, up to 20 bar, the effect of steam addition (8 bar partial pressure), and the effect of particle size (up to 1.6 mm) on BOF slag carbonation extent. Fig. 4 presents results on the effect of three CO₂ pressures (4, 12 and 20 bar) on the carbonation of the 0.08-0.5 mm sieved fraction of BOF_{II} slag, performed at three temperatures (350, 500 and 650 °C) for 30 minutes reaction duration. These sets of conditions were chosen to ensure the effects of the process parameters could be observed (i.e. carbonation for longer duration or at higher temperature would mask the effect of pressure). It was found that the effect of CO₂ pressure on CO₂ uptake (reported as wt% CO₂ in the carbonated sample, determined by in-situ calcination and IR CO₂ detection) is more important at lower temperatures, having a significant effect at 350 °C (176% improvement at 20 bar over 4 bar), a much reduced effect at 500 °C (7% improvement), and no effect at 650 °C. The effect of temperature, at fixed CO₂ pressures, was more striking, as had been presented with the in-sitt TGA carbonation methodology. At 4 bar CO₂ pressure there was a 400% improvement in CO₂ uptake at 650 °C over 350 °C; significant improvements were also measured at 12 bar (+130%) and 20 bar (+80%).

Fig. 5 presents pressurized basket reactor carbonation results on the effect of slag particle size for both types of BOF slag (BOF_I and BOF_{II}). Furthermore these experiments tested the effect of steam addition (8 bar partial pressure) on CO_2 uptake. The experimental temperature was fixed at 650 °C, the best value from the previous set of experiments presented, reaction time was 30 minutes, and total pressure was maintained at 20 bar ($CO_2 + H_2O$ where applicable). Steam had a positive effect on CO_2 uptake for all particle size fractions, though the effect was reduced for the largest particles. Furthermore on a percent improvement basis, BOF_I benefited more from steam addition than the more reactive (due to greater free

lime content) BOF_{II}. For reference, pure CaO was also carbonated in the pressurized basket reactor at 650 °C and 20 bar total pressure with and without steam. Steam addition allowed carbonation conversion to reach 100% of the theoretical value, compared to 64% in dry CO₂.

The effect of steam on lime carbonation has previously been reported in literature by Yang and Xiao [39] and Manovic and Anthony [38]; however the mechanism of action has not yet been fully characterized. Yang and Xiao [39] suggest steam introduces a 'catalytic effect', related to the adsorption of hydroxyl groups and the formation of bicarbonate intermediates, though no experimental data is provided to confirm this hypothesis. They claim Ca(OH)₂ is not likely formed since carbonation temperatures in their tests were above the decomposition temperature of Ca(OH)₂ at the utilized vapor pressures of water. Indeed, it would appear difficult to base the effect of steam on the hydration of free lime, given experimental conditions for hot-stage carbonation normally use temperatures beyond the decomposition temperature of calcium hydroxide at trace vapor pressure of water (i.e. condition within a TGA using inert carrier gas) of approximately 350-400 °C. However, in the present experiment the vapor pressure of water was 8 bar, which applying the model of Hartman and Martinovský [41] yields a decomposition temperature (dehydroxylation reaction $\Delta G = 0$) of 660 °C. Therefore it is plausible to assert that the presently observed steam-enhanced CO₂ uptakes may be related to an intermediate hydroxylation step prior to carbonation. This would be in agreement with reported results of Materic and Smedley [42], who found that hydration of lime prior to atmospheric carbonation can result in greater carbonation conversion when applying lime as a sorbent in calcium looping processes for CO₂ capture and concentration.

Manovic and Anthony [38], on the other hand, provide experimental evidence that the steam-enhancement of carbonation may be caused by improvement in solid state diffusion rather than hydroxylation. They show that steam does not improve carbonation kinetics during the initial fast (kinetically controlled) carbonation stage, but rather aids in carbonation extent improvement over prolonged time (i.e. during the diffusion controlled stage). The exact mechanism of how steam improves solid state diffusion, however, is not provided. In any case, regardless of the actual mechanism of steam assistance, the present experiments show that steam improvement of carbonation conversion of the slags is

modest (i.e. coarse particles remain distant from theoretical maximum uptake). Therefore industrial application of steaming during hot-stage carbonation of BOF slag could be deemed optional, depending on process complexity and required or desired final product properties.

It is evident from Fig. 5 that particle size, and hence surface area, has a significant effect on CO₂ uptake. This is not surprising since direct carbonation is a surface based and diffusion controlled reaction. Moreover, BOF₁ CO₂ uptake (Fig. 5a) was consistently lower than BOF_{II}, which is attributable to its lower free lime content (10.6 wt% versus 20.6 wt%). Converting CO₂ uptake into carbonation conversion of the free lime content of the slags (i.e. assuming, based on experimental evidence collected in this study, that free lime is the only mineral participating in BOF slag direct high-temperature carbonation), the difference between BOF_I and BOF_{II} diminishes (Fig. 5b). In fact, based on the shrinking core model, it would be expected that carbonation conversion of the two slag samples would be identical if: (i) solid state diffusivity of CO₂ in the two slags were equal; (ii) free lime was equally distributed within the particles of both slags (radially and concentrically); and (iii) the proportion of free lime present in each slag as precipitated free lime and residual free lime were equal. Fig. 5b indicates that the two finest fractions of BOF_{II} maintain an advantage with regards to carbonation conversion over BOF_I. This can be attributed to the moderately higher specific surface areas and lower average particle sizes of the two finest fractions of BOF_{II} compared to BOF_I (Table 2), and may also be due to differences in dispersion and morphology of free lime inclusions (precipitated and residual) in the two slags (for example, precipitated free lime may carbonate more/better than the residual variety). For the two coarser fractions, dry carbonation conversions were more similar, as it can be expected that carbonation of these samples is mainly controlled by solid state diffusion rather than their limited (not measured) surface areas. Still, the average percentage improvement in steam-assisted carbonation conversion for the two coarse fractions was similar to that of the two finest fractions (+46% and +40%, respectively). Therefore improvement of solid state diffusivity as proposed by Manovic and Anthony [38] appears to be a plausible explanation of the mechanism, though hydroxylation or another 'catalytic' effect cannot be entirely ruled out.

3.2.3 Atmospheric Furnace Carbonation

Subsequent to the two previously described experimental methods (in-situ TGA carbonation and pressurized basket reactor carbonation), atmospheric furnace carbonation experiments were conducted for two main reasons. Firstly, atmospheric carbonation appears to be a more feasible route for industrial implementation of hot-stage carbonation, as it appears from previous tests that pressure is not a critical parameter for satisfactory direct carbonation extent, and it should reduce technical complexity and thus processing cost. Secondly, with regards to the laboratory scale equipment used in this study, this methodology allowed production of larger quantities (up to 10 g) of carbonated material at one time (i.e. identical processing conditions), enabling detailed study on the mineralogy, basicity and heavy metal leaching properties of the carbonated products.

Similar to the previous methodologies, the effect of temperature on CO₂ uptake was studied to ensure furnace carbonation delivered equivalent results based on previous findings, and to elucidate unforeseen effects at atmospheric pressure. Furthermore, the effect of reaction time was studied, varied between 10 minutes (in-situ TGA carbonation indicated rapid CO₂ uptake in the first 10 minutes of reaction time) and 6 hours (longest reaction time tested thus far, to ensure cessation of CO₂ uptake). Fig. 6a shows that, in what appears to be the optimal temperature range of 650 to 750 °C, there is only a small increase in carbonation extent when increasing the reaction time from 10 minutes to 6 hours. At lower temperatures, such as 500 °C, the effect of time is more evident, but below this temperature, especially in the range of 200 to 400 °C, the carbonation extent is much smaller. Above 750 °C the stability of CaCO₃ decreases due to the low CO₂ pressure (Table 5), and at 800 °C the carbonation extent significantly decreases with 10 minutes reaction time (possibly the furnace is not completely purged of inert gas within 10 minutes of CO₂ introduction, hence the CO₂ partial pressure is somewhat lower than 1 atm, compromising CaCO₃ stability at this temperature). Results are very similar for both batches of BOF used, the difference being BOF_{II}, which contained more free lime, achieved greater CO₂ uptake amounts.

Quantitative XRD was also used to assess the carbonation extent of atmospheric furnace carbonated BOF_{I} slag samples (Fig. 6b). These analyses were conducted on unmilled 0.08-0.5 mm particle size fraction samples (i.e. not micronized to 5-10 μ m as normally done for compositional analysis using XRD). The aim was to limit X-ray penetration into the particle to determine the conversion of free lime closer to

the particle surface, which is the region that controls material basicity and heavy metal leaching. It is estimated that the X-ray penetration depth is limited to tens of microns up to 100 microns (depending on X-ray source and target material) [43], which is to some extent less than average particle radius based on D_{50} and D[4,3] values (Table 2). By comparing the calcite (CaCO₃) amounts to the remaining free lime (CaO) and portlandite (Ca(OH)₂) amounts, it was possible to determine carbonation conversion of the free lime present in the region penetrated by the X-rays. Values between 50-60% conversions are obtained in the optimal temperature range (600 to 750 °C), slightly higher than overall free lime conversions based on total CO_2 uptake (40-50%). Given that the carbonation depth may be smaller than the X-ray penetration depth, and/or that CaO grains are not fully carbonated due to formation of dense carbonated shell surrounding the unreacted core, detection by XRD of free lime present in regions of the particles that are not accessible to CO_2 is possible, therefore truly geometrical peripheral near-surface carbonation extent is likely to be higher.

The effect of particle size on CO₂ uptake by atmospheric furnace carbonation (Fig. 7a) was analogous to results obtained by pressurized basket reactor carbonation (Fig. 5a). The 1.0-1.6 mm particle size fraction of BOF₁ slag achieved significantly lower CO₂ uptake than the finer 0.08-0.5 mm fraction after one hour carbonation at 700 °C. The reduction in basicity of the coarse samples, however, was very similar to the reduction obtained with the finer samples (Fig. 7b). This is explained by the fact that carbonation primarily occurs near the particle surface, which contacts the liquid phase and controls pH. Therefore reduction in basicity is not always directly proportional to CO₂ uptake, but rather depends on particle size and most importantly on surface area.

3.3 Basicity and Heavy Metal Leaching

The native basicity of BOF slag can be directly linked to its free lime content, which is detrimental to its valorization as a construction material. Furthermore, materials with high basicity when in contact with water produce alkaline solutions, which contribute to the leaching of several heavy metal contaminants [32,44,45]. For a specific slag particle size fraction, CO₂ uptake and basicity reduction are proportionally

linked. Atmospheric furnace carbonation resulted in improvement of solution pH values of the 0.08-0.5 mm particle size fraction of BOF_I slag; the basicity reduction was proportional to CO₂ uptake, as presented in Fig. 8. BOF_I, having less free lime, achieved just under one pH unit reduction with 10 minutes reaction time (Fig. 8a), and reached incrementally better results after 6 hours reaction time (Fig. 8b). BOF_{II} benefitted more from increased reaction time, probably due to its significantly greater free lime content and moderately higher specific surface area. Nevertheless, at the optimal conditions (e.g. 700°C) the final pH values for BOF_I and BOF_{II} after 10min/6hr carbonation were 11.6/11.6 and 12.1/11.7 respectively. This reduction in pH may be sufficient to increase the marketability of BOF slag into applications in the construction industry; however it remains to be tested whether swelling propensity sufficiently decreases given the incomplete free lime conversions achieved.

Positive effects on the leaching of heavy metals also resulted from carbonation of BOF slags.

Leaching results as a function of atmospheric-furnace-carbonated sample pH are plotted in Fig. 9. For further insight, leaching results as a function of CO₂ uptake and reaction time are illustrated in Figs. SC-11 and SC-12 in the Supplementary Content, the former for 10 minutes reaction time, and the latter for 6 hours reaction time. The 0.08-0.5 mm particle size fraction carbonated samples were used to generate these data.

Carbonation was found to have a noticeable positive effect on reducing leaching of barium, nickel and cobalt. These results can be attributed to pH reduction (Fig. 9), which decreases the solubility and/or changes the speciation of some metals, and formation of less soluble compounds (e.g. BaCO₃, BaSO₄ [44]). Vanadium leaching, on the other hand, increases after carbonation, reaching 10 mg/kg (well above the regulatory limit of 0.8 mg/kg [32]) for three of the four sample sets. The only exception is 10 minutes carbonated BOF_{II} slag (Fig. SC-11), which remains below the limit. This can be explained by the lower reduction in basicity of this material (Fig. 8a), as vanadium is known to be more prone to solubilization at lower pH [15,45]. Chromium and molybdenum leaching also increase after carbonation (pH reduction), the former more than the latter; however their values remain below regulatory limits (Table 1). Chromium leaching, like vanadium, is also proportional to basicity reduction [45], as inferred from the 10 minutes carbonated BOF_{II} slag presenting marginal leaching increase (Fig. SC-11) compared to 6 hours reaction

time (Fig. SC-12). Competition of carbonate ions with oxyanionic species can also play a role in controlling the solubility of chromate, molybdate and vanadate ions [32].

It can be concluded that the balance of basicity reduction and heavy metal leaching via carbonation is a challenge for the reutilization of BOF slag. While the solubility of some metals reduced after carbonation (barium, cobalt and nickel), vanadium and chromium leaching increased. With more moderate carbonation extent (10 minutes reaction time), the rate of solubility increase of these metals was lower, conforming to regulatory limits. Therefore, in view of industrial implementation of BOF slag carbonation, it should be considered whether a higher or lower carbonation extent is more interesting for commercialization of carbonated BOF slag. The increased solubility of vanadium (and to a smaller extent chromium) after carbonation can be used as a means to ease their recovery from the residues, in comparison to energy intensive traditional methods for vanadium recovery from waste materials [46], as these metals can be of commercial value. Alternatively, if the carbonated BOF slag is used as a fine aggregate in cement mortar/concrete, there are possibilities to introduce additives to the matrix (such as zeolitic sorbents) to contain leaching from the aggregate material. Moreover, the matrix itself can act as a sink for heavy metals, by incorporation of metalloids including Cr and V in hydration products [47], thereby suppressing leaching.

4. Conclusions

This work has investigated the hot-stage carbonation of Basic Oxygen Furnace (BOF) steel slag, with the objective of converting its high free lime content (10-20 wt%) to calcium carbonate and stabilizing its negative properties (high basicity and heavy metal leaching) that prevent material valorization. By performing hot-to-cold carbonation experiments in a TGA, it was observed that the optimum carbonation temperature lies just below the transition temperature between carbonation stability and carbonate decomposition: 830-850 °C at 1 atm CO₂, and 750-770 °C at 0.2 atm CO₂. Pressurization and steam addition were found not to contribute significantly to CO₂ uptake enhancement; for instance, diffusion limitations prevent free lime conversion from surpassing 50% for the 0.08-0.5 mm particle size fraction of

the BOF_{II} slag batch. Instead, it appears feasible to apply flue gases from steel production plants directly (e.g. 0.2 atm CO₂ partial pressure) without need for additional and costly CO₂ separation/concentration processes. Improvement in basicity was independent of particle size; that is, even the coarser fractions achieved similar reduction in basicity despite significantly lower CO₂ uptake. However, the effect of carbonation on heavy metal leaching was mixed: while the solubility of some metals reduced after carbonation (barium, cobalt and nickel), vanadium and chromium leaching increased. Post-carbonation extraction of leachable components or mixing of carbonated slag with sorbent materials would be required for meeting hazardous materials regulations and enable slag commercialization. Otherwise, heavy metal leaching could be suppressed by applying carbonated BOF slag as a fine aggregate in cement mortar/concrete, where the encasing matrix could act as a sink for mobile metals.

Further optimization of hot-stage carbonation is still required prior to industrial implementation of this technology. Despite suitable reaction kinetics and enhanced CO₂ diffusivity at high temperatures, complete conversion of free lime is still limited by the ability of CO₂ to diffuse into the particle core, even for the 0.08-0.5 mm particle size fraction utilized. This size fraction is at the low-end of the expected particle size achieved by hot-stage granulation (as envisaged in Fig. SC-1 and reported in [10,25]). Therefore in order to achieve reasonably high levels of free lime conversion, development of granulation/hot-stage-milling technology capable of delivering sub-millimeter sized particles appears to be essential. Furthermore, investigation is warranted on finding the necessary level of free lime conversion needed to satisfactory stabilize BOF swelling in construction material applications. For instance, it should be assessed whether 50% conversion or more is required for complete stabilization, or if near-surface carbonation, which is enough to reduce basicity (as achieved in this work with 1.0-1.6 mm BOF particle size fraction), may be sufficient to improve material properties and enable its valorization.

Acknowledgements

The K.U.Leuven Industrial Research Fund is gratefully acknowledged for funding the Knowledge Platform on Sustainable Materialization of Residues from Thermal Processes into Products (SMaRT-Pro²)

in which this work was performed. R.M.S is thankful for financial support from the Natural Sciences and Engineering Research Council of Canada (NSERC) by a PGS-D scholarship, and from the Research Foundation - Flanders (FWO) by an international mobility grant. Special regards to Marc Nolmans for assistance with the tube furnace.

References

- [1] Y. Topkaya, N. Sevinç, A. Günaydın, Slag treatment at Kardemir integrated iron and steel works, Int. J. Miner. Process. 74 (2004) 31–39.
- [2] R. Dippenaar, Industrial uses of slag (the use and re-use of iron and steelmaking slags), Ironmaking Steelmaking 32 (2005) 35–46.
- [3] J.-M. Delbecq, Steel Slags as cementitious materials, in: Seminário Internacional Aplicação de Escória de Aciaria, Belo Horizonte, Brazil, 2010.
- [4] G. Wang, Y. Wang, Z. Gao, Use of steel slag as a granular material: Volume expansion prediction and usability criteria, J. Hazard. Mater. 184 (2010) 555–560.
- [5] J.J. Emery, Slag Utilization in Pavement Construction, Extending Aggregate Resources, ASTM Spec. Tech. Publ. 774 (1982) 95–118.
- [6] S.A. Mikhail, A.M. Turcotte, Thermal behaviour of basic oxygen furnace waste slag, Thermochim. Acta 263 (1995) 87–94.
- [7] H. Motz, J. Geiseler, 2001, Products of steel slags an opportunity to save natural resources, Waste Manage. 21 (2001) 285–293.
- [8] K. Toshiyuki, K. Akio, Treatment of Cooling Steelmaking Slag, Japan Patent, JP-06-184610, 05 July 1994.
- [9] S. Chengxiao, L. Yongqian, W. Jiangang, L. Yin, Technology of slag processing by rotary cylinder at Baosteel, Rev. Metall. 1 (2004) 39–42.
- [10] O. Sang-Yoon, Method for Stabilizing Slag and Novel Materials Produced Thereby, US Patent, US2009/0193849, 06 Aug 2009.

- [11] K.S. Lackner, D.P. Butt, C.H. Wendt, Progress on binding CO₂ in mineral substrates, Energy Convers. Manage. 38, (1997) S259–S264.
- [12] W.J.J. Huijgen, R.N.J. Comans, Carbonation of Steel Slag for CO₂ Sequestration: Leaching of Products and Reaction Mechanisms. Environmental Sci. Technol. 40 (2006) 2790–2796.
- [13] W.J.J. Huijgen, G.-J. Witkamp, R.N.J. Comans, Mineral CO₂ Sequestration by Steel Slag Carbonation, Environ. Sci. Technol. 39 (2005) 9676–9682.
- [14] E.-E. Chang, C.-H. Chen, Y.-H. Chen, S.-Y. Pan, P.-C. Chiang, Performance evaluation for carbonation of steel-making slags in a slurry reactor, J. Hazard. Mater. 186 (2011) 558–564.
- [15] A. van Zomeren, S.R. van der Laan, H.B.A. Kobesen, W.J.J. Huijgen, R.N.J. Comans, Changes in mineralogical and leaching properties of converter steel slag resulting from accelerated carbonation at low CO₂ pressure, Waste Manage. 31 (2011) 2236–2244.
- [16] T. Isoo, T. Takahashi, M. Fukuhara, Using Carbonated Steelmaking Slag Blocks to Help Reduce CO₂, Am. Ceram. Soc. Bull. 80 (2001) 73–75.
- [17] S. Eloneva, S. Teir, H. Revitzer, J. Salminen, A. Said, C.-J. Fogelholm, R. Zevenhoven, Reduction of CO₂ Emissions from Steel Plants by Using Steelmaking Slags for Production of Marketable Calcium Carbonate, Steel Res. Int. 80 (2009) 415–421.
- [18] R.M. Santos, T. Van Gerven, Process Intensification Routes for Mineral Carbonation, Greenhouse Gas Sci. Technol. 1 (2011) 287–293.
- [19] E.P. Reddy, P.G. Smirniotis, High-Temperature Sorbents for CO₂ Made of Alkali Metals Doped on CaO Supports, J. Phys. Chem. B 108 (2004) 7794–7800.
- [20] J. Blamey, E.J. Anthony, J. Wang, P.S. Fennell, The calcium looping cycle for large-scale CO₂ capture, Prog. Energy Combust. Sci. 36 (2010) 260–279.
- [21] V. Manovic, E.J. Anthony, Lime-Based Sorbents for High-Temperature CO₂ Capture—A Review of Sorbent Modification Methods, Int. J. Environ. Res. Public Health 7 (2010) 3129–3140.
- [22] V. Prigiobbe, A. Polettini, R. Baciocchi, Gas–solid carbonation kinetics of Air Pollution Control residues for CO₂ storage, Chem. Eng. J. 148 (2009) 270–278.

- [23] L.-C. Kao, Sorption of Carbon Dioxide at High Temperatures by Slag, Master's Thesis, National Cheng Kung University, Tainan City, Taiwan, 2009.
- [24] J. Yu, K. Wang, Study on Characteristics of Steel Slag for CO₂ Capture, Energy Fuels 25 (2011) 5483–5492.
- [25] G. Li, H. Ni, Recent progress of hot stage processing for steelmaking slags in China considering stability and heat recovery, in: Proceedings of the Second International Slag Valorisation Symposium: The transition to sustainable materials management, Leuven, Belgium, 2011.
- [26] F. Larachi, I. Daldoul, G. Beaudoin, Fixation of CO₂ by chrysotile in low-pressure dry and moist carbonation: Ex-situ and in-situ characterizations, Geochim. Cosmochim. Acta 74 (2010) 3051–3075.
- [27] S.K. Bhatia, D.D. Perlmutter, Effect of the Product Layer on the Kinetics of the CO₂-Lime Reaction, AIChE J. 29 (1983) 79–86.
- [28] D. Gielen, CO₂ removal in the iron and steel industry, Energy Convers. Manage. 44 (2003) 1027–1037.
- [29] A. Tobiessen, Process Evaluations and Simulations of CO₂ Capture from Steel Plant Flue Gases, in: CCS in the Iron and Steel Industry: Challenges and Opportunities of CO₂ Capture and Storage in the Iron and Steel Industry, Dusseldorf, Germany, 2011.
- [30] F. Larachi, J.-P. Gravel, B.P.A. Grandjean, G. Beaudoin, Role of steam, hydrogen and pretreatment in chrysotile gas—solid carbonation: Opportunities for pre-combustion CO₂ capture, Int. J. Greenhouse Gas Control 6 (2012) 69–76.
- [31] Ministere de la Region Wallonne, Arrêté du Gouvernement wallon favorisant la valorisation de certains déchets, Moniteur Belge 2 (2001) 23859–23883.
- [32] T. Van Gerven, E. Van Keer, S. Arickx, M. Jaspers, G. Wauters, C. Vandecasteele, Carbonation of MSWI-bottom ash to decrease heavy metal leaching, in view of recycling, Waste Manage. 25 (2005) 291–300.
- [33] K. Mohan, F.P. Glasser, The thermal decomposition of Ca₃SiO₅ at temperatures below 1250°C I. Pure C₃S and the influence of excess CaO or Ca₂SiO₄, Cem. Concr. Res. 7 (1977) 1–7.

- [34] F. Wachsmuth, J. Geiseler, W. Fix, K. Koch, K. Schwerdtfeger, Contribution to the Structure of BOF-Slags and its Influence on Their Volume Stability, Can. Metall. Q. 20 (1981) 279–284.
- [35] J. Waligora, D. Bulteel, P. Degrugilliers, D. Damidot, J.L. Potdevin, M. Measson, Chemical and mineralogical characterizations of LD converter steel slags: A multi-analytical techniques approach, Mater. Charact. 61 (2010) 39–48.
- [36] R.M. Santos, D. François, G. Mertens, J. Elsen, T. Van Gerven, Ultrasound-Intensified Mineral Carbonation, Appl. Therm. Eng. (2012) doi:10.1016/j.applthermaleng.2012.03.035.
- [37] M. Schmidt, Mechanical and thermal carbonation of strontium ferrite SrFeO_x, Mater. Res. Bull. 37 (2002) 2093–2105.
- [38] V. Manovic, E.J. Anthony, Carbonation of CaO-Based Sorbents Enhanced by Steam Addition, Ind. Eng. Chem. Res. 49 (2010b) 9105–9110.
- [39] S. Yang, Y. Xiao, Steam Catalysis in CaO Carbonation under Low Steam Partial Pressure, Ind. Eng. Chem. Res. 47 (2008) 4043–4048.
- [40] C. Chen, S.-T. Yang, W.-S. Ahn, Calcium oxide as high temperature CO₂ sorbent: Effect of textural properties, Mater. Lett. 75 (2012) 140–142.
- [41] M. Hartman, A. Martinovský, Thermal stability of the magnesian and calcareous compounds for desulfurization processes, Chem. Eng. Commun. 111 (1992) 149–160.
- [42] V. Materic, S.I. Smedley, High Temperature Carbonation of Ca(OH)₂, Ind. Eng. Chem. Res. 50 (2011) 5927–5932.
- [43] J. Liu, R.E. Saw, Y.-H. Kiang, Calculation of Effective Penetration Depth in X-Ray Diffraction for Pharmaceutical Solids, J. Pharm. Sci. 99 (2010) 3807–3814.
- [44] A.-M. Fällman, Leaching of chromium and barium from steel slag in laboratory and field tests a solubility controlled process?, Waste Manage. 20 (2000) 149–154.
- [45] G. Cornelis, A. Johnson, T. Van Gerven, C. Vandecasteele, Leaching mechanisms of oxyanionic metalloid and metal species in alkaline solid wastes: A review, Appl. Geochem. 23 (2008) 955–976.
- [46] G.Z. Ye, Vanadium Recovery from LD Slag, A State of the art report, Part I Facts and metallurgy of Vanadium, Internal MEFOS report, MEF06001, Stockholm, Sweden, 2006.

[47] N. Saikia, G. Cornelis, G. Mertens, J. Elsen, K. Van Balen, T. Van Gerven, C. Vandecasteele, Assessment of Pb-slag, MSWI bottom ash and boiler and fly ash for using as a fine aggregate in ACCEPTED MANUSCRIP cement mortar, J. Hazard. Mater. 154 (2008) 766-777.

List of Figures

- **Fig. 1.** Major mineral phases of BOF_I slag identified via SEM-BSE-EDX.
- **Fig. 2.** In-situ TGA carbonation of pure oxides (CaO, MgO and Fe₂O₃) and micronized slags (BOF_I and BOF_{II}): hot-to-cold (subfigures a and b) and cold-to-hot (subfigure c) methodologies; $P_{CO2} = 1$ atm, except where indicated.
- **Fig. 3.** In-situ TGA carbonation of micronized slags (BOF_I (a) and BOF_{II} (b)) and pure CaO (c): fixed temperature (500, 575, 650, 725 and 800 $^{\circ}$ C) methodology; $P_{CO2} = 1$ atm.
- **Fig. 4.** Pressurized basket reactor carbonation of 0.08-0.5 mm particle size fraction of BOF_{II} slag at multiple temperatures (350, 500 and 650 $^{\circ}$ C) and CO_2 pressures (4, 12 and 20 bar) for 30 minutes reaction time.
- **Fig. 5.** Pressurized basket reactor carbonation of BOF_1 and BOF_{II} slag samples of varying particle size fractions (< 0.08, 0.08-0.5, 0.5-1.0 and 1.0-1.6 mm) at 650 °C with dry or wet CO_2 at 20 bar total pressure and 30 minutes reaction time: results expressed as (a) wt% CO_2 uptake, and (b) carbonation conversion of free lime content.
- **Fig. 6.** Atmospheric furnace carbonation results: (a) CO₂ uptake (determined by TGA analysis) and (b) peripheral near-surface free lime conversion (determined by unmilled XRD analysis with Rietveld refinement) for carbonation of 0.08-0.5 mm particle size fractions of BOF_I and BOF_{II} slags at varying fixed temperatures (200 to 800 °C) for 10 minutes and 6 hours reaction time.
- **Fig. 7.** Atmospheric furnace carbonation results: (a) CO₂ uptake and (b) basicity for carbonation of two particle size fractions (0.08-0.5, 1.0-1.6 mm) of BOF₁ slag at 700 °C for 1 hour reaction time.

Fig. 8. Atmospheric furnace carbonation results: basicity of 0.08-0.5 mm particle size fractions of BOF_I and BOF_{II} slags as a function of CO_2 uptake for 10 minutes (a) and 6 hours (b) carbonation reaction time. Varying CO_2 uptakes resultant from carbonation at multiple temperatures (i.e. same samples presented in Fig. 6a).

Fig. 9. Atmospheric furnace carbonation results: leaching of heavy metals (Ba, Ni, V (left-side); Cr, Co, Mo (right-side)) from 0.08-0.5 mm particle size fractions of BOF_I (top) and BOF_{II} (bottom) slags as a function of pH. Varying pH values resultant from carbonation at multiple temperatures and reaction times (i.e. same samples presented in Fig. 6a).

List of Tables

Table 1Heavy metal leaching limits for waste material re-use in Belgium [31,32].

	V	Cr	Co	Ni	Mo	Ba
Limit ^c (mg/kg)	0.80 a	1.00 b	1.00 b	2.00 b	1.50 b	1.60 a

^a Flemish/Dutch: NEN7343; ^b Walloon: DIN 38414/EN 12457-4/CEN TC 292.

 $^{^{}c}$ Walloon testing procedure uses batch extraction test at L/S = 10; Flemish testing procedure uses column leaching test at L/S = 10.

Table 2Average particle sizes of prepared BOF slag fractions determined by laser diffraction and specific surface areas determined by nitrogen adsorption (BET).

Slag batch	BOF_I	BOF_{II}	BOF _I	BOF _{II}	BOF_{I}	BOF _{II}
-						
Particle size				0.00		0.00
	micronized	micronized	< 0.08 mm	< 0.08 mm	0.08-0.5 mm	0.08-0.5 mm
range						
D[4,3] (μm)	13.6	13.5	38.0	31.0	252.5	203.9
					(1)	
$D[3,2] (\mu m)$	1.7	1.3	2.9	2.5	68.4	49.7
D (um)	10.7	7.3	30.5	23.5	224.7	179.5
D_{50} (μm)	10.7	1.5	30.3	23.3	224.7	119.5
S_{BET} (m ² /g)	8.2	10.5	5.4	7.8	1.6	1.7
22. (0)						

Table 3Elemental composition of BOF slag batches determined by XRF (measured as oxides, presented as elemental).

Elements	Ca	Fe	Si	Mn	Al	Μα	Ti	Cr	V	Balance
(wt%)	Ca	re	51	1 V111	Ai	ivig	11	Cı	V	(H, C, O, trace)
BOF _I slag	35.3	21.1	5.5	2.7	1.1	0.52	0.40	0.32	0.17	32.9
BOF _{II} slag	41.3	15.8	5.9	2.9	0.62	0.33	0.30	0.12	0.15	32.6

Table 4 Mineral composition (wt%) of BOF slag batches determined by XRD with Rietveld refinement (normalized to 100% of the identified crystalline phases).

Mineral composition (wt%) of BOF slag batches determined by XRD w						
(normalized to 100%	of the identified crysta	lline phases).				
Mineral	Chemical formula	BOF _I	BOF _{II}			
Srebrodolskite (C ₂ F)	Ca ₂ Fe ₂ O ₅	37.2	31.0			
β -C ₂ S	Ca ₂ SiO ₄	21.5	22.0			
Lime	CaO	10.6	20.6			
γ -C ₂ S	Ca ₂ SiO ₄	7.3	7.9			
Fayalite	Fe ₂ SiO ₄	5.5	4.2			
Hematite	Fe_2O_3	4.1	2.0			
Enstatite	$MgSiO_3$	3.8	1.9			
Ferrosilite	FeSiO ₃	2.8	3.6			
Quartz	SiO ₂	2.8	1.4			
Wollastonite	CaSiO ₃	1.1	0.7			
Wüstite	FeO	1.1	2.2			
Magnetite	Fe ₃ O ₄	0.8	0.4			
Periclase	MgO	0.7	1.6			
Iron	Fe	0.4	0.3			

Table 5Carbonation/calcination equilibrium temperatures (Gibbs free energy of reaction equal to zero) for alkaline mineral components of BOF slag (FactSage 6.2 modeling).

Mineral	Reaction Eq.	P(CO ₂) (atm)	$T (\Delta G = 0) (^{\circ}C)$
CaO	(1)	0.0003	511
CaO	(1)	0.01	627
CaO	(1)	0.1	732
CaO	(1)	1	861
CaO	(1)	5	971
Ca ₂ SiO ₄	(2)	1	400
$Ca_2Fe_2O_5$	(3)	1	664
MgO	(4)	1	400

- Novel hot-stage carbonation approach to Basic Oxygen Furnace (BOF) slag presented.
- Optimum carbonation rate at boundary of equilibrium (830-850 °C at 1 atm CO₂).
- Limited effect of pressure (0.2 to 20 bar CO₂) and steam (8 bar H₂O) on CO₂ uptake.
- rient.
 Co and Ni. Slag CO₂ uptake proportional to particle size, surface area and free lime content.

























