

Remediation of acid mine drainage and immobilization of rare earth elements: Comparison between natural and residual alkaline materials

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ARTICLE INFO

Keywords:

Acid mine water remediation
Passive treatment systems
Waste products
Blast furnace slag
Biomass ash
Rare earth elements

ABSTRACT

Acid mine drainage (AMD) is a well-known source of toxic trace metals in freshwaters. Traditional passive treatment systems rely on AMD neutralization with limestone and removal of most common toxic transition metals such as Cu and Zn with little attention to rare earth elements (REE). Alkaline waste materials now receive increasing attention as low cost AMD treatment alternatives in the circular economy. This study was set up to identify the efficiency of alkaline waste materials remediating AMD and scavenging REE in addition to other toxic trace elements. An AMD sample was collected from a lixiviante coming from pyrite heaps in the Iberian Pyrite Belt (pH = 1.8, 30 μM ΣREE). The sample was treated with either blast furnace slag (BFS) generated during smelting of iron ore in a blast furnace or biomass ashes (BA) derived from combustion of biomass, thereby using analytical grade CaCO_3 and NaOH as reference products. The batch alkalization experiments were conducted by adding each alkaline material at an amount to obtain an equal pH to ≈ 6.5 . The required amounts of the products were $\text{NaOH} < \text{CaCO}_3 < \text{BFS} < \text{BA}$ in line with their acid neutralizing capacities. The largest removal of sulfate from water was obtained in the CaCO_3 treatment suggesting gypsum precipitation which was lower with BA and BFS and virtually absent with NaOH, these trends were confirmed by SEM-EDX and XRD. Both BFS and BA removed more Fe than CaCO_3 and NaOH. The REE elements were well removed by all treatments (>99%) and the remaining REE concentrations in the solutions were clearly lower than values for Cu and Zn. The Zn and Cu removals were not consistently high enough (except with NaOH) to meet environmental limits in the discharge waters. The largest efficiency for REE removals was obtained with CaCO_3 . Indirect evidence here suggests that gypsum is a better host for the trivalent REE than Fe(III) minerals in the precipitates. The ionic radii of trivalent REE are more similar to Ca^{2+} than to Fe^{3+} , explaining the better potential of gypsum as REE host. This study showed also the potential of BFS as alkaline agent for the remediation of AMD in terms of its higher alkalinity generation potential as compared to BA, thus making BA less promising than BFS.

1. Introduction

Acid mine drainage (AMD) is a major source of metal contamination of freshwaters. Its generation is mainly induced by the oxidation of sulphidic ores i.e. pyrite (FeS_2) during mining and mineral exploitations. This acid flow then facilitates dissolution of elements contained in sulfides as impurities such as Cu, Zn, Cd, As and Pb. In addition, the high acidity released also leads to the dissolution of other elements, i.e. Al, Ca, Mg, Na, K and rare earth elements (REE) contained in the host rocks

surrounding the sulfides. The low pH increases the solubility and mobility of REE, which are commonly found in AMDs in concentrations several orders of magnitude higher than those found in natural freshwater bodies (Ayora et al., 2016; Noack et al., 2014; Wang et al., 2021; Zänker et al., 2003; León et al., 2021). If left untreated, AMD may contaminate aquatic and terrestrial environments even after closing of the mining activities (Doye and Duchesne, 2003; Akinwekomi et al., 2016; Duraes et al., 2017; Ferreira da Silva et al., 2009; Masindi, 2017; Perez-Lopez et al., 2011; Ruehl and Hiibel, 2020; Sheoran, 2006;

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<https://doi.org/10.1016/j.apgeochem.2023.105800>

Received 11 July 2023; Received in revised form 21 September 2023; Accepted 26 September 2023

Available online 27 September 2023

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Totsche et al., 2006; Turingan et al., 2022). The historical mining area stretching from southwest of Spain to Portugal known as the Iberian Pyrite Belt (IPB) is one example illustrating the severity of AMD generation. The IPB contains one of the largest concentrations of massive sulfide deposits in the world. More than 100 abandoned mines present in the IPB have been collectively contributing to this pollution problem for decades (Duraes et al., 2017; Gonzalez et al., 2020; Orden et al., 2021; Soyol-Erdene et al., 2018).

The AMD can be treated by using either active treatment systems (ATS) or passive treatment systems (PTS) (Ben Ali et al., 2019). ATS, which involve a water treatment plant, are a prerequisite of active mining operations which have to comply a strict control of the final effluent quality before being released in the environment (Turingan et al., 2022). PTS are generally used for the rehabilitation of closed and abandoned mine sites. ATS are based on the continuous addition of chemicals (i.e., NaOH) and use of electricity (Macías et al., 2017; Skousen et al., 2019; Cánovas et al., 2020), while PTS do not require this continuous addition and employ naturally available energy sources i.e., gravity flow and natural oxidation. Natural material, such as limestone, or even constructed wetlands (Orden et al., 2021; Caraballo et al., 2009; Hengen et al., 2014; Macias et al., 2012a) are used in PTS for remediation. The PTS require, therefore, less maintenance and lower energy costs than ATS (Perez-Lopez et al., 2011; Ruehl and Hiibel, 2020; Turingan et al., 2022; Orden et al., 2021; Bernier, 2005; Ness et al., 2014; Sheridan et al., 2013). Some PTS, such as the Dispersed Alkaline Substrate (DAS) technology which is in use at the Mine Esperanza and Mine Concepción in the IPB (Orden et al., 2021) rely on the pH neutralization and subsequent metal precipitation using alkaline materials with limestone (CaCO_3) mainly for trivalent metal precipitation (i.e. Fe^{3+} and Al^{3+}) and complemented by an additional treatment step with MgO, for divalent metal precipitation (i.e. Cu^{2+} , Zn^{2+} , Mn^{2+} etc.) (Caraballo et al., 2009). This neutralization is performed in tanks filled with a mixture of coarse materials (i.e., wood chips) and alkaline materials to prevent the clogging problems related to Fe precipitation commonly observed in PTS applied to AMD (Orden et al., 2021). Inside the limestone-DAS PTS, AMD flows down in a vertical direction and gets in contact with CaCO_3 . Upon contact, CaCO_3 starts to dissolve and generates alkalinity (Lozano et al., 2019a). Along this alkalization process, between pH 3–4.5, Fe starts to precipitate as poorly crystalline iron oxyhydroxysulphate (i.e. schwertmannite ($\text{Fe}_8\text{O}_8(\text{SO}_4)_x(\text{OH})_y \cdot n\text{H}_2\text{O}$, where $x = 1.4\text{--}1.5$, $y = 5.0\text{--}5.2$)) (Lozano et al., 2019a). Then, when the pH further increases to pH 4.5–6, Al starts to precipitate. Lozano et al. (2020) and Sánchez-España et al. (2011) identified microcrystalline aluminium oxyhydroxysulphate (i.e. basaluminite ($\text{Al}_4(\text{SO}_4)(\text{OH})_{10} \cdot 15\text{H}_2\text{O}$)) as the main formed Al phase (Lozano et al., 2020a; Sánchez-España et al., 2011). Ferrihydrite, a low crystallinity Fe (III) oxide ($\text{Fe}(\text{OH})_3$) is also a possible precipitate above pH > 5.5 and some amorphous forms of $\text{Al}(\text{OH})_3$ (i.e. Gibbsite) could also precipitate above pH 5 (Lozano et al., 2019a, 2019b, 2020a; Sánchez-España et al., 2011). Manganese containing mineral phases, such as birnessite ($(\text{Na}, \text{Ca})\text{Mn}_7\text{O}_{14}$) and/or todorokite ($(\text{Ca}, \text{Na}, \text{K})_x\text{Mn}_6\text{O}_{12}$) were also observed to precipitate along the alkalization process (Hedin et al., 2019). Gypsum ($\text{CaSO}_4 \cdot 2\text{H}_2\text{O}$) may precipitate regardless of the pH change along the alkalization process due to the high concentrations of sulfate and Ca in the pore waters after calcite (CaCO_3) dissolution (Ferreira da Silva et al., 2009; Lozano et al., 2020a; Sánchez-España et al., 2011; Lozano et al., 2019b; Bortnikova et al., 2020). Schwertmannite, basaluminite and gypsum have been identified as the major secondary minerals precipitated inside the limestone-DAS PTS (Orden et al., 2021; Rotting et al., 2008). At the end, the treated AMD flowing out from such limestone-DAS PTS in the IPB has a pH value of around 6.5 and almost 100% removal of trivalent metals has been observed (Noack et al., 2014; Orden et al., 2021). Though divalent metals precipitation is favored above pH 8, the precipitation of Fe and Al containing secondary mineral phases have been shown to scavenge transition metals and arsenate (Ayora et al., 2016; Orden et al., 2021; Sánchez-España et al., 2011;

Lozano et al., 2019b).

The AMDs can contain REE at concentration levels ranging between 4000 and 80,000 pmol/L (Ayora et al., 2016). The REE possess unique properties which make them technology critical elements (Ferreira da Silva et al., 2009; Ayora et al., 2013; Hassas et al., 2020; Royer-Lavallee et al., 2020; Los Elementos de las Tierras, 2015; Bau et al., 2018; Costis et al., 2021; Protano FR, 2002; Migaszewski and Galuszka, 2014) and due to increased emissions into the environment (Ferreira da Silva et al., 2009; Perez-Lopez et al., 2011; Gonzalez et al., 2020; Soyol-Erdene et al., 2018; Costis et al., 2021; Protano FR, 2002; Migaszewski and Galuszka, 2014; Gammons et al., 2005; Lozano et al., 2020b; Park et al., 2016; Prudencio et al., 2015; Sahoo et al., 2012; Sikakwe et al., 2018; Stolpe et al., 2013), REE are also considered as emerging contaminants. Hence scavenging them from AMD in a PTS for further recycling may be attractive, both to limit emissions in the environment and to provide more sustainable REE sources. Today there is a rather limited attention to the risks and mitigations of REE in AMD, likely because the environmental risk assessments of REE are still in their infancy. Studies showed that some PTS scavenged REE (Costis et al., 2021; Prudencio et al., 2015; Naidu et al., 2019). Several removal mechanisms, such as co-precipitation and/or sorption (i.e. surface complexation) can be responsible for this REE scavenge along the PTS (Ferreira da Silva et al., 2009; Lozano et al., 2020a; Sánchez-España et al., 2011; Lozano et al., 2019b). Though the number of studies remains limited, mapping techniques, such as synchrotron based μ -X-ray fluorescence (μ XRF) and field emission scanning electron microscopy with energy-dispersive X-ray spectroscopy (FESEM-EDS) showed the potential retention of REE by Al-precipitates (i.e. basaluminite) in these systems (Ayora et al., 2016; Hedin et al., 2019).

There is an increasing attention in using alkaline waste materials in PTS to replace limestone, as it could improve the environmental footprint. Those materials represent economically and environmentally promising alternatives (Jones and Cetin, 2017; Millán-Becerro et al., 2021). Their use enables waste recycling leading to industrial symbiosis (Jones and Cetin, 2017). Nevertheless, it is important to achieve similar or better removal of contaminants present in AMD compared to the classical reagents for both heavy metals and REE. Examples of such waste materials are blast furnace slag (BFS) generated during smelting of iron ore in a blast furnace, and biomass ashes (BA) derived from combustion of biomass. The global annual production of BA is estimated to be 480 Mt and the bulk of it, as a primary management strategy, ends up in landfills (Millán-Becerro et al., 2021; Bogush et al., 2019). The global annual production of BFS exceeds 25 million Mt (Dzięcioł and Radziemska, 2022). Slag is usually used in road construction industry as an aggregate, however managing huge amounts of slag still remains a challenge (Dzięcioł and Radziemska, 2022). BFS and BA contain oxides that are not only a source of alkalinity, such as CaO, MgO and K_2O but also contain amorphous and crystalline phases and thus reactive surfaces to which REE and transition metals can adsorb, i.e. these waste products may be superior alkaline materials compared to lime only, a hypothesis that still remains to be tested. Given the potential of BFS and BA to be used as alkaline materials in remediating AMD will allow recycling of those waste materials and hence handling of huge waste piles will no longer represent a problem. Moreover, it is important to better understand the mechanisms involved in REE removal using these waste materials. This would allow, if ever judged as an interesting prospect, to better design REE retrieval methods from PTS.

This study was set up to identify the efficiency of alkaline waste materials in remediating AMD and scavenging REE. The strategy was to compare the removal of REE and trace elements as a function of the BFS or BA dose with corresponding doses of limestone or NaOH, all at an equivalent near neutral pH (about 6.5). The CaCO_3 , BFS and BA all contain Ca and will likely induce gypsum precipitation. To identify the effect of gypsum, pure NaOH was included as a reference material not containing Ca. The waste material and the formed secondary mineral phases were characterized by SEM-EDX, XRD and/or XRF to get some

insight into the REE immobilization mechanisms.

2. Materials and methods

2.1. Acid mine drainage sampling and characterization

The AMD used in the experiments was collected from a permanent leachate flowing out from a sulfide dump located in the NE of the Filon Norte open pit in the Tharsis mining area. The Tharsis mining area is located in the Huelva province (SW Spain) and is the second mine in importance of the IPB. This area is one of the most polluted in the IPB (Tornos et al.; González et al., 2022; Moreno González et al., 2019; Santisteban et al., 2019; Valente et al., 2013) and most interesting for the purpose of the present study due to its high concentration of REE (Moreno González et al., 2019). The selected AMD is representative of extremely acid and metal-rich mining leachates of the Iberian Pyrite Belt and are characterized by seasonal variations with maximum concentrations of elements typically present in AMD in summer (dry period) and due to evaporation and lower ones in winter (wet period) due to dilution effect by rainfall. More details about seasonal variations of acid mine drainage from the Tharsis mines are given by Moreno-González et al. (2022) (Gonzalez et al., 2020). The AMD sample was collected at the end of April which is a representative of an intermediate period (between wet and dry period).

Field measurements were carried out at the sample location with a HACH sensION™ + MM150 portable multimeter probe to determine pH, electrical conductivity (EC), redox potential (ORP) and temperature. Before use, the electrodes were calibrated according to the manufacturer's instructions using certified materials. The AMD water sample was collected in pre-washed (10% HNO₃) polyethylene containers. An aliquot of the sample was filtered through a 0.22 μm PVDF membrane filter unit. A fraction was acidified with HNO₃ Suprapur to prevent metal precipitation and kept at 4 °C until analysis by Inductively Coupled Plasma Mass Spectrometry (ICP-MS) (Agilent 7700x ICP-MS) for its elemental composition (Fe, Al, transition metals and REE). A second filtered fraction was kept at 4 °C without acidification for subsequent anion analysis using Ion Chromatography IC (IC Dionex) and total organic and inorganic carbon content analysis using a TIC/TOC analyzer (Shimadzu TOC-L-CPH analyzer). The ultrapure water produced with a Milli-Q water purification system (Sartorius Water purific. SAR10) was used for necessary sample dilutions before any analysis. For the quality control of the method, blank samples, reference standards and duplicate analysis were used.

2.2. Alkaline waste materials

The BFS used in this study was obtained in powdered form from Ecocem Benelux B.V. (Netherlands). It is generated by smelting of iron ore in a blast furnace and subsequently dried and ground into a fine powder. The BA was obtained from a small power plant (50 MW) located in Huelva in Spain and operated by ENCE where over 500,000 tons of biomass is burnt annually for electricity production. BA was sieved to obtain homogenized sample (<63 μm). The two solids were characterized using different techniques including X-ray Fluorescence (XRF), Scanning Electron Microscopy coupled with an Energy Dispersive Spectroscopy (SEM/EDX) and X-ray diffraction (XRD). A Spectro XEPOS HE XRF was used to determine the elemental composition of BFS and BA. The analyses were performed on dried fine grinded (<250 μm) samples (±10 g each). A Phenom Desktop SEM equipped with an EDX device was used to observe the morphology and to identify the elemental composition of the waste materials. Before analysis, the solids were pressed into 1.3 cm sized pellets, mounted on aluminum stubs and coated with an ultrathin layer (5 nm) of gold to make their surfaces conductive. Imaging was done at a voltage of 10 kV, while EDX spot analyses targeting major phases were performed at an elevated voltage of 15 kV. The presence of crystalline phases in BFS and BA were

determined by using XRD (Bruker D8 Advance) equipped with a Cu Kα source, a beam knife, a Ni filter and automated divergence slits. Operational settings were set at 40 kV and 40 mA and measurements were done between 5 and 60°2θ at a step size of 0.02°2θ. With the aim of quantitatively analyzing the mineral phases present, the powdered samples were mixed with 10 wt% of ZnO as internal standard. XRD measurements were then obtained using thin films of powder samples. Data semi-quantification was done using the Rietveld refinement method and the content of amorphous phases in each sample was calculated based on the use of ZnO as internal standard.

The acid neutralizing capacity (ANC) of BFS and BA was determined using pH titration. The definition of ANC depends on the selected final pH (Wahlström et al., 2009). Here, about 0.1–0.5 g of each waste material was mixed with 100 ml MilliQ water (initial pH of the solutions were around 9.6–9.9) and subsequently titrated with 0.1 M HCl (prepared by diluting Titrisol®) down to pH 6.5, each in duplicates. This pH was selected as a representative to what is usually achieved in limestone-DAS PTS in IPB. The ANC was calculated according to Equation (1).

$$\text{Acid neutralizing capacity (ANC), mmol} / \text{kg} = \frac{\text{mmol H}^+}{\text{material, kg}} \quad \text{Equation 1}$$

With mmol H⁺, the amount of H⁺ added and the material in kg, the mass of solid titrated.

2.3. Batch alkalization experiments

Preliminary batch alkalization experiments were performed by adding different amounts of BA, BFS, analytical grade NaOH or CaCO₃ to the AMD solution to determine the quantity of each material to reach pH of ~6.5. In general, the intention of installing PTS is to achieve neutral pH in the treated water to fulfill the requirement imposed by the Food and Agriculture Organization of the United Nations (FAO) for irrigation waters. However, the PTS used in the Spanish part of the IPB being based on the DAS system are able to increase the pH only up to ≈ 6.5. Therefore, it was decided to raise the pH to ≈6.5 to compare the efficiencies of BFS and BA with CaCO₃. The solutions were shaken on an orbital shaker (Ohaus Orbital Shaker) under a constant speed for two weeks and pH was recorded. Based on that, final experiments were set up by mixing the required doses of alkaline materials (Table 1) to 20 mL AMD solution in 50 mL Nalgene tubes to raise the pH to ~6.5, each in duplicate. Suspensions after the addition of the alkaline materials were kept in the orbital shaker under a constant speed for two weeks to ensure proper contact between the added alkaline materials and the AMD solution. Based on the preliminary batch studies, this contact period was sufficient to achieve a stable pH. The precipitated solids and supernatants were subsequently collected after separation by centrifugation (Sigma 6-16 KS) at 7000×g for 10 min and the pH of the supernatants were measured using InLab Routine Pro probe (Mettler Toledo). The supernatants were then filtered at 0.22 μm on PVDF filter. A portion of the filtrates was acidified with 1% (V/V) HNO₃ (65% Suprapur, Merck) to avoid metal precipitation and analyzed by ICP-MS to obtain the final concentrations of Fe, Al and trace elements including REE. Another portion of the filtrates was analyzed by IC to obtain the final concentrations of sulfate.

The solids were freeze-dried (Telstar LyoQuest 85 Plus) before

Table 1
Required dose (alkaline material:AMD) ratio to increase pH to ~6.5 and the resulting pH.

Alkaline material	pH	Dose (g/L)
NaOH	6.43 ± 0.05	15 ± 0.18
CaCO ₃	6.47 ± 0.10	23 ± 0.10
BFS	6.67 ± 0.05	36 ± 0.03
BA	6.32 ± 0.10	185 ± 0.30

characterization. The morphology and the elemental analysis of the formed solid samples were characterized by SEM/EDX (Phenom Desktop) following the same sample preparation and analysis procedure as mentioned before. The identification and (semi)-quantification of the formed secondary phases were examined by XRD as described previously.

2.4. Thermodynamic modelling

The geochemical code PHREEQC (Parkhurst and Appelo, 2013) Interactive (version 3.6.2–15100) together with the thermodynamic database Thermodem_v1.10 (Blanc et al., 2012) were used to calculate the saturation state of possible mineral phases in the AMD before and after alkalization and the REE speciation in solution. The Thermodem_v1.10 database (the activity model utilized is Bdot (Parkhurst and Appelo, 1999)) was chosen as it contains thermodynamic data for all REE. The database was complemented by the incorporation of thermodynamic data for the mineral phases schwertmannite and basaluminite found in literature. The respective equilibrium reactions and constants are shown in Table SI 1 in the supporting information.

According to Wood et al. (1990), REE di-sulfate complexes (i.e. $\text{Ln}(\text{SO}_4)_2$) become important when the sulfate concentration in solution containing REE is greater than 0.02 M (Wood, 1990). Therefore, the REE speciation of the AMD was performed after incorporating $\text{Ln}(\text{SO}_4)_2$ complexes in the ThermodemV1.10 (TD) thermodynamic database (TDB) (Blanc et al., 2012). The thermodynamic data of the $\text{Ln}(\text{SO}_4)_2$ complexes (except for Ce) were integrated into the TDB from the Lawrence Livermore National Laboratory (LLNL) TDB (Spahiu and Bruno, 1995) by performing the required thermodynamic data conversions to keep consistency.

3. Results and discussion

3.1. Acid mine water

The pH of the AMD measured in the field was 1.8, confirming the severity of AMD pollution around the sample collection area. The low pH level is related to the extreme sulfuric acid concentrations (0.4 mol/L) (Table 2). The ORP of 397 mV indicates the oxidizing nature of this AMD and the EC value of 18.5 mS/cm reveals the presence of high concentrations of ions which is further confirmed by analysis of the AMD (Table 2). In addition to the high concentration of sulfate, the high concentrations of Fe, Al and Mn together with relatively high concentrations of other trace metals (i.e. Zn, Cu, Co, Ni, etc.) make AMD very toxic justifying the requirement for remediation (Ayora et al., 2016;

Table 2

The concentration (<0.22 μm) of major, transition elements and REE measured by ICP-MS, SO_4^{2-} measured by IC and TIC/TOC measured by TIC/TOC analyzer (Means and standard deviation; n = 2).

Major elements and transition elements		Rare earth and yttrium (REY) elements	
mmol/L		$\mu\text{mol/L}$	
SO_4^{2-}	400 ± 20	La	2 ± 0.1
Mg	100 ± 5.1	Ce	6 ± 0.3
Fe	70 ± 3.2	Pr	0.8 ± 0.06
Al	40 ± 2.7	Nd	4 ± 0.2
Ca	10 ± 0.4	Sm	1 ± 0.06
Na	10 ± 0.6	Eu	0.3 ± 0.02
Zn	7 ± 0.3	Gd	2 ± 0.04
Mn	4 ± 0.1	Tb	0.2 ± 0.01
Cu	1 ± 0.6	Dy	1 ± 0.07
Co	0.2 ± 0.01	Y	10 ± 0.0
Ni	0.1 ± 0.01	Ho	0.2 ± 0.01
As	0.05 ± 0.0	Er	0.7 ± 0.03
TIC	0.02 ± 0.003	Tm	0.09 ± 0.00
TOC	0.2 ± 0.07	Yb	0.5 ± 0.03
		Lu	0.07 ± 0.01

Perez-Lopez et al., 2011; Lozano et al., 2020b; Reguer et al., 2020). The high concentrations of REY ($\sum \text{REY} = 29 \mu\text{mol/L}$) (Table 2) also confirms their increased mobilization in AMD related to the sulfide-sulfate oxidation processes.

To put it into perspective, compared to the respective limit values recommended by the FAO for irrigation waters (Ayers and Westcott, 1985), the concentrations of Fe, Al, Mn, Cu, Zn and sulfates in this AMD are 811, 266, 1146, 574, 233 and 65 times higher, respectively. Therefore, impelled firstly by the European Water Framework Directive (Moreno-Gonzalez et al., 2022), and the urgent need of water for agricultural activities, there is a plan to remediate the catchment around this area (Lozano et al., 2020b).

The concentrations of REE in AMD normalized to European Shale (EUS) show an upward convex curvature shaped pattern suggesting Middle REE (MREE) enrichment in this AMD (Fig. 1). In fact, the ratios of Er_N/Gd_N and Gd_N/La_N were < 1 and > 1, respectively which confirmed this enrichment. The MREE enrichment is a common feature of such AMD (Ayora et al., 2016; Prudencio et al., 2015; Sahoo et al., 2012) and is due to the preferential release of MREE from MREE enriched minerals such as monazite, apatite (e.g. phosphates) and bastnasite (e.g. carbonates in the host rock under acidic conditions. (Prudencio et al., 2015; León et al., 2023).

Geochemical modelling of saturation states revealed undersaturation of schwertmannite upon four different K_s commonly used in literature (schwertmannite1-4: see Table SI 2 in the supporting information), as well as ferrihydrite, basaluminite and gibbsite in the AMD before alkalization. The SI of Gypsum was however close to zero, meaning thus close to equilibrium (Table SI 2). Therefore, gypsum seems to control aqueous sulfate concentration in the AMD.

The calculated REE speciation in the AMD is shown in Figure SI 1. It is clear that REE mono-sulfate complexes (i.e. LnSO_4^+) are the dominating species in the AMD for most of the REE. Yet, REE di-sulfate complexes $\text{Ln}(\text{SO}_4)_2$ are also shown to be important species in this AMD in agreement with the 0.02 M sulfate concentration threshold stated by Wood (1990). Moreover, small percentages of free REE species (i.e. Ln^{3+}) are also predicted to be present in the AMD.

Further modelling about the alkalization experiments was done and will be discussed in the respective result and discussion section (under section 3.3).

3.2. Alkaline materials

The elemental composition of BFS and BA obtained from X-ray fluorescence analysis are presented in Table 3.

The BFS is mostly composed of Ca and Si containing phases. According to the literature, BFS usually has a CaO/SiO_2 ratio of

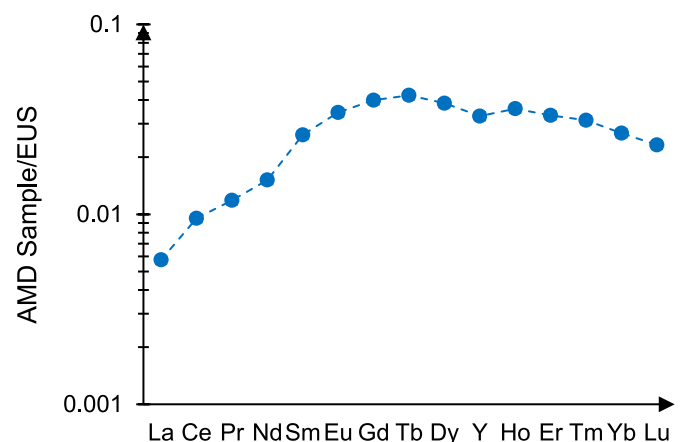


Fig. 1. Normalized REE concentrations (<0.22 μm) with respect to European Shale (Means and standard deviation; n = 3).

Table 3
Percent oxide composition of BFS and BA using XRF.

%	Na ₂ O	MgO	Al ₂ O ₃	SiO ₂	K ₂ O	CaO	Fe ₂ O ₃	TiO ₂
BFS	0.8	8.6	9.9	38.5	0.4	40.6	0.5	0.6
BA	3.1	3.9	11.3	53.8	4.4	18.3	4.4	0.7

approximately 1, which is consistent with the value measured here, i.e. 1.1 (Simmons et al., 2014). Moreover, nearly 95% of the total slag is usually composed primarily of SiO₂, Al₂O₃, CaO and MgO, which is also the case here (97%) (Dzięcioł and Radziemska, 2022; Simmons et al., 2014). BFS is usually defined as a calcium aluminosilicate rich waste due to presence of this combination of elements and oxides (Fayed et al., 2021). BA is mostly composed of Si followed by Ca which is consistent with the XRD mineralogical information mentioned below (Table 5). Unlike BFS, BA usually has a variable composition (Bogush et al., 2019). The Vassilev's chemical classification mentioned in (Bogush et al., 2019) can be used to identify the type of ash and its level of acid tendency (the ability to release H⁺ ions in aqueous solution), which is useful to evaluate the efficiency of ashes to remediate AMD (Bogush et al., 2019). According to that classification, BA used in this study belongs to the siliceous CaO rich ash with medium acid tendency (SC-MA type) which is in fact good for AMD remediation (Bogush et al., 2019). This is confirmed by the ANC of BFS and BA at pH 6.5 shown in Table 4. The ANC of BFS and BA mainly comes from the dissolution of Ca, Mg, K and Na contained in the (amorphous) phases therein (Table 3). The BFS has ~2 times higher ANC than BA, which is consistent with the higher content of basic oxides in BFS (~50%) as compared to BA (~30%).

The XRD results presented in Table 5 show that BFS is mainly composed of amorphous mineral phases and small quantities of calcite and merwinite [Ca₃Mg(SiO₄)₂]. The BA is mainly comprised of amorphous phases and quartz. In addition, plagioclase feldspar [(Na,Ca)(Si, Al)₄O₈] and calcite were also identified in BA. The respective XRD spectra are shown in the supplementary information (Figure SI 3). Calcite is present in both BFS and BA and is thus contributing to the alkaline generating capacity. Merwinite which is a commonly found mineral in slag comprises also Ca and Mg and upon dissolution at acidic pH may also contribute to the alkaline generating capacity of BFS (Engström et al., 2013). However, this effect can be considered minor/negligible as the present merwinite content in BFS is very small compared to the other phases therein. This is also true for plagioclase feldspar in BA (OXBURGH et al., 1994). In addition, these amorphous and crystalline phases could serve as sorbing surfaces for REE and transition metals.

The morphology of BFS and BA examined using SEM and by their elemental composition measured using EDX are shown in Fig. 2. The SEM images show that BFS and BA mainly consist of irregular-shaped particles with diameters up to 30 μm. The EDX confirms the total elemental analyses: BFS mainly composes of Ca, Si, Al, Mg as major elements with some S as impurity. The presence of oxygen in the EDX results indicates the presence of oxides as well. Silica in BFS is present as amorphous SiO₂ rather than quartz, which is not present based on the XRD analysis. The BA consists of Si, Ca, Al, K and Fe as major elements with Mg and Na as impurities.

3.3. Batch alkalization experiments

The doses required to raise the pH of AMD to pH = 6.5 ranked BA >

Table 4
Acid neutralizing capacity of BFS and BA (Means and standard deviation; n = 2).

	ANC down to pH 6.5, mmol H ⁺ /kg
BFS	504 ± 3.0
BA	264 ± 4.2

Table 5
Semi-quantitative XRD analysis of BFS and BA.

Wt %	Merwinite [Ca ₃ Mg(SiO ₄) ₂]	Calcite [CaCO ₃]	Plagioclase (Na, Ca) feldspar [(Na, Ca)(Si,Al) ₄ O ₈]	Quartz [SiO ₂]	Amorphous
BFS	0.7	0.3	0	0	99
BA	0	2.6	3.5	24	70

BFS > CaCO₃ > NaOH (Table 1) and are in line with the inverse of the ANC of these products (Table 4). The theoretical ANC to pH 6.5 of CaCO₃ and NaOH are 20,000 mmol/kg (or 10,000 mmol/kg if only dissolves to HCO₃⁻ and no CO₂ escapes) and 25,000 mmol/kg, respectively.

The % removal of sulfate after the alkalization experiments with the four different alkaline materials is shown in Table 6. The most effective sulfate removal is achieved by using CaCO₃, while the least effective sulfate removal is observed when using NaOH. This result is consistent with the Ca content of the alkaline materials and their capacity to induce gypsum precipitation. The BFS and BA remove equal fractions of sulfate even though they do not have the same Ca content.

According to the FAO standards for irrigation waters (Ayers and Westcot, 1985), the residual sulfate concentration should not exceed 5.2 × 10⁻³ mol/L. None of the alkaline materials was able to reach that standard because none of these have a sufficiently soluble Ca. This was also the case after the limestone-DAS PTS at the Mine Esperanza in the IPB in which a residual sulfate concentration of 2.1 × 10⁻² mol/L (around a 30% sulfate removal) was present (Orden et al., 2021). This implies the need for additional treatment for removal sulfates. Studies showed that the use of BaCO₃-DAS system as an additional unit after the limestone-DAS was promising in this regard, i.e. increased the sulfate removal via the precipitation of BaSO₄ (Gomez-Arias et al., 2015; Torres et al., 2018). Alternatively, speciation calculations show that more sulfate can be precipitated as gypsum using CaCl₂ as a source of soluble Ca²⁺.

It should be noted that, however, a direct comparison is not possible with the limestone-DAS system installed in the field because of the difference between the CaCO₃ that was used in this study and the natural limestone that is being used in the field (i.e. DAS system together with woodchips). Though the matrixes are different, the trends obtained from this study are similar to the ones in the "real" limestone-DAS system.

The batch alkalization experiments were simulated by using the initial AMD composition (Table 2) and experimental doses of CaCO₃ and NaOH (Table 1) as reactants with gypsum added as a secondary phase as it is the dominating phase precipitation during AMD alkalization. Fe- and Al- phases (schwertmannite and basaluminte) precipitation were not taken into account as at this stage it is not certain which type of schwertmannite or basaluminte have been precipitated and thus having no constraint for the simulations. Alkalization path was represented by hundred calculation steps and the reaction step that matched best with the measured pH was used to estimate the amount of gypsum precipitation. This was done with the intention of obtaining rough calculation to assess gypsum precipitation. However, more detailed modelling will be done in the future considering already compiled thermodynamic data of Fe- and Al- phases (schwertmannite and basaluminte).

The alkalization of the AMD/pH is mainly related to CaCO₃ dissolution (related to H⁺ consumption) (Table SI 3) while the precipitation of Fe- and Al- oxyhydroxysulfates are associated to H⁺ release (Table SI 3). Gypsum precipitation is also contributing to H⁺ consumption (Table SI 3). Therefore, the precipitation of Fe- and Al- oxyhydroxysulfates may play a role in buffering the pH of the system and also in the sulfate precipitation during AMD alkalization.

Batch reaction calculations predict that CaCO₃ addition to reach pH 6.5, results in gypsum precipitation (~1.8 × 10⁻¹ mol/L). Mass balance calculations reveal that however ~50% (about 1.8 × 10⁻¹ mol/L) of sulfate still remain in solution. This amount is slightly higher than the measured residual sulfate (Table 6). In the experiments using NaOH, the

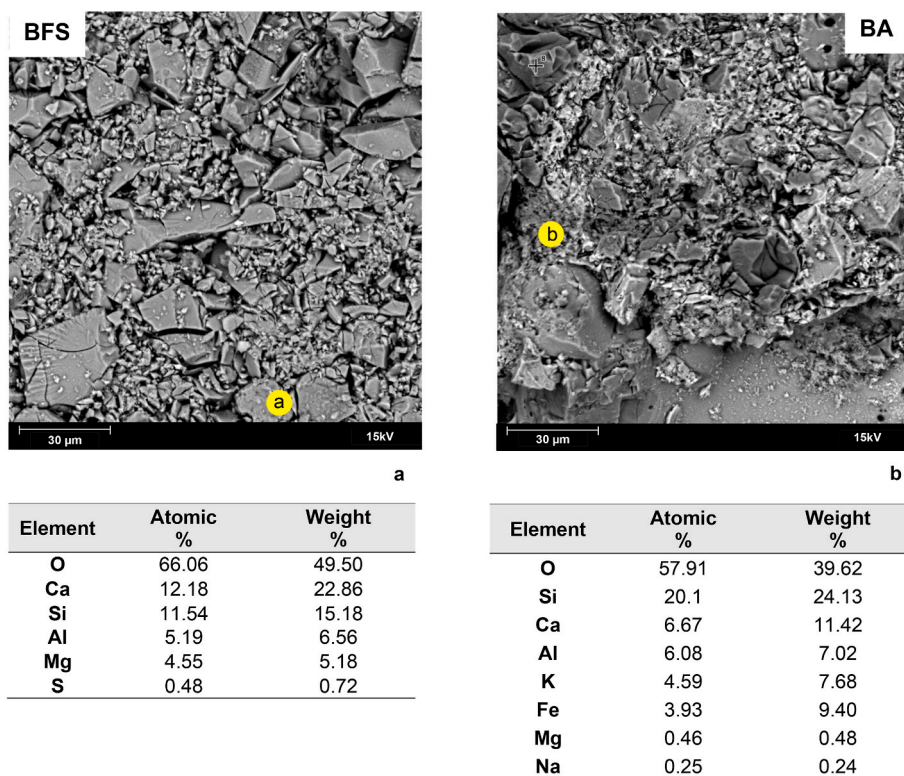


Fig. 2. SEM images of BFS and BA with the elemental composition as measured by EDX.

Table 6

residual sulfate concentration and % removal of sulfate after the alkalization (Means and standard deviation; n = 2).

	Residual sulfate concentration, mol/L	Removal of sulfate [%]
NaOH	0.27 ± 0.03	19 ± 4
CaCO ₃	0.13 ± 0.03	62 ± 3
BFS	0.16 ± 0.02	54 ± 2
BA	0.16 ± 0.01	54 ± 2

geochemical calculations predict precipitation of only 3.1×10^{-3} mol/L of gypsum, logically leading to a higher residual sulfate concentration (about 3.6×10^{-1} mol/L), which is slightly higher than the measured one (Table 6). In both cases the slightly higher calculated residual sulfate concentrations can be attributed to the fact that sulfate containing Fe- and Al-phases were not considered in the calculations. No geochemical calculations were done for BFS and BA, as their exact compositions are not known and no thermodynamic data are available for these materials. The reactions referred to in the text above (occurring during alkalization) are given in Table SI 3.

The XRD analysis showed that the precipitated solids after the alkalization experiments mainly consist of amorphous phases, irrespective of the alkaline material used (Table 7). The respective XRD spectra are provided in the supplementary information (Figure SI 4).

Table 7

(Mineral) composition of the precipitates after the experiment using XRD characterization.

Alkaline agent	wt%			
	Amorphous	Gypsum	Quartz	Calcite
NaOH	100	–	–	–
CaCO ₃	72	18	–	9
BFS	64	36	–	–
BA	76	3	21	–

Only amorphous phases precipitated in the NaOH treated solutions. The XRD results clearly show gypsum precipitation during CaCO₃, BFS and BA alkalization which is attributed to the Ca in the added alkaline materials. The lowest sulfate removal obtained with NaOH (Table 6) can therefore be explained by the absence of gypsum precipitation during this alkalization as indicated by the XRD analysis. The other crystalline phases observed by XRD after BA and CaCO₃ alkalization, i.e., quartz and calcite, respectively, correspond most likely to the primary quartz and calcite present in the unreacted materials.

The SEM images of the samples after alkalization using the four different alkaline materials are shown in Fig. 3. The Ca, S and O rich phases correspond to gypsum minerals with characteristic elongated shapes and are visible in the treatments with CaCO₃ (denoted by the spots of d and f), with BFS (denoted by the spots of h and i) and with BA (denoted by the spots of k and m) but are not visible with NaOH. These results are in agreement with XRD results (Table 7). Moreover, Fe, Al, S and O rich phases which are not clearly crystalline were common with all four different alkaline materials: NaOH (denoted by the spots of a, b and c), CaCO₃ (denoted by the spots of e and g), BFS (denoted by the spot of j) and BA (denoted by the spots of n and l). Therefore, the amorphous phases identified by XRD should mainly be composed of Fe, Al, S and O rich phases i.e., Fe and Al-containing oxyhydroxysulfates. The spots of a, b, c, e, g, j, l and n contain trace amounts of Zn while the spots of b, e and n contains trace amounts of Cu as well. This indicates Zn and Cu removal along with those Fe, Al, S and O rich phases precipitation. None of the REE-concentrations was detectable with SEM/EDX.

3.3.1. Metal removal

The residual concentrations of Al, Mn, Fe, Cu and Zn remaining in solution after the alkalization are shown in Fig. 4.

The lowest residual Al concentration of 2.7×10^{-5} mol/L is obtained after alkalization with BA, whereas the highest residual Al concentration of 8.0×10^{-5} mol/L is measured after alkalization with NaOH. The residual Al concentrations using CaCO₃ and BFS are similar, i.e., 4.0×10^{-5} mol/L. Thus, BA yields a better Al removal than BFS. Overall, all

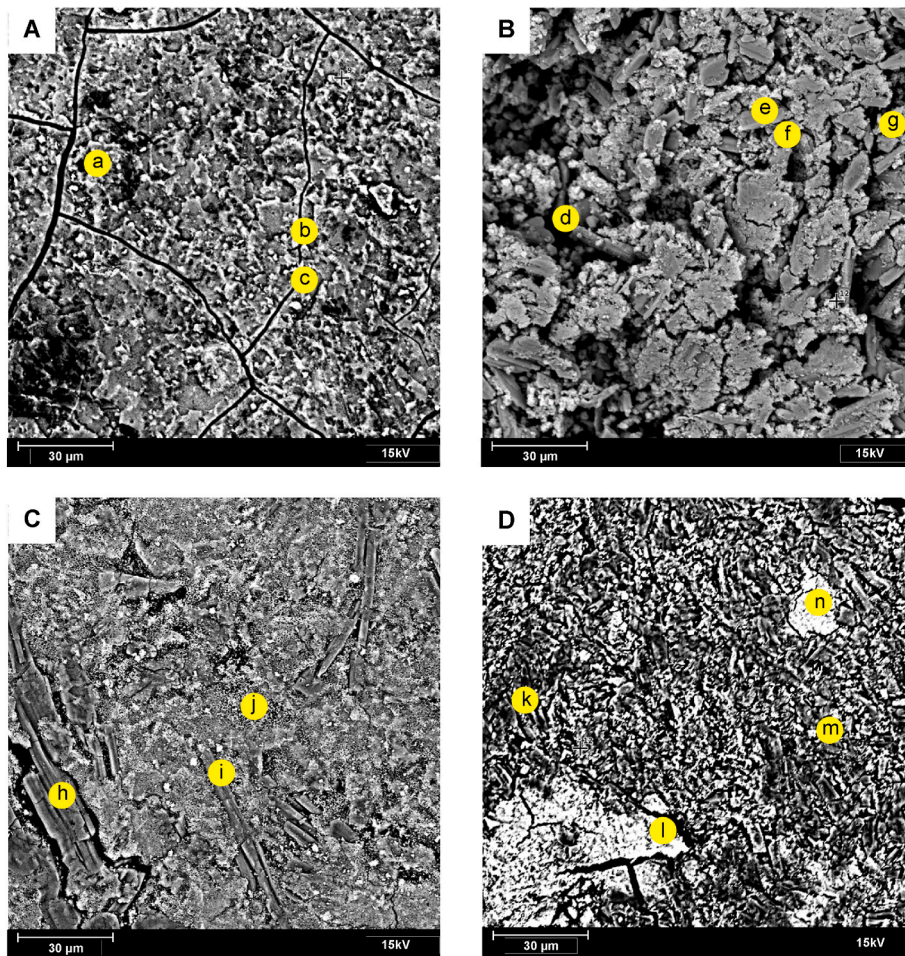


Fig. 3. SEM images of samples after the alkalization experiments with NaOH: A; CaCO₃: B; BFS: C and BA: D.

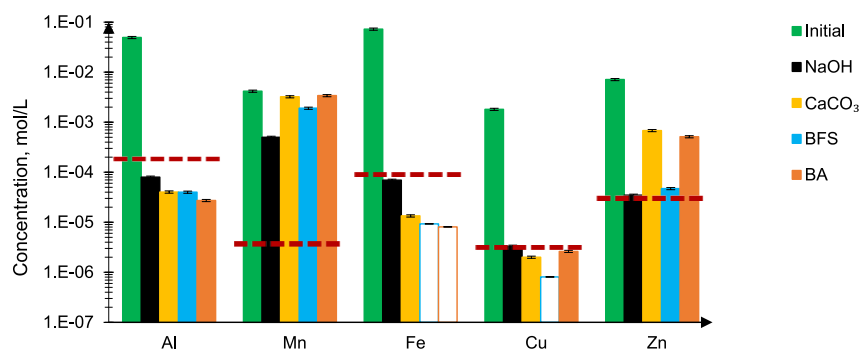


Fig. 4. Initial and residual concentrations of Al, Mn, Fe, Cu and Zn. (Dashed lines indicate the respective FAO irrigation waters standards (Ayers and Westcot, 1985); Unfilled bars: detection limit. Measured concentrations were below that limit. Means and standard deviation (n = 2 after treatment and n = 3 before treatment).

four materials show above 99.8% removal of Al after the alkalization which represents an efficient remediation of the initial AMD. Moreover, alkalization using all four materials were able to meet the criteria of the FAO for irrigation water. According to literature, for the treated AMD flowing out of the limestone-DAS PTS at the Mine Esperanza abandoned mine, a residual concentration of 3.7×10^{-6} mol/L was achieved (Orden et al., 2021). This is better than what is obtained after CaCO₃ alkalization during our experiments. Yet, it is noteworthy that

the PTS at the Esperanza Mine consists of two successive limestone-DAS reactors to improve its performances (Orden et al., 2021).

The residual concentration of Fe after the alkalization with NaOH and CaCO₃ corresponds respectively to 6.9×10^{-5} mol/L and 1.3×10^{-5} mol/L. The residual Fe concentration with BFS is less than 9.3×10^{-6} mol/L, while with BA it is less than 8.1×10^{-8} mol/L, which accounts for more than 99.9% Fe removal. The Al and Fe comprised in the materials did not increase the concentration of these elements in

solution, the residual concentrations of these two elements were even lower in BA and BFS treatments compared to CaCO_3 and NaOH. Similar to Al, alkalization using all four materials were also able to comply with the recommended value given in the FAO for irrigation water for Fe (i.e., 8.95×10^{-5} mol/L). Fe removal efficiency during the first limestone DAS unit at the Esperanza Mine was only 49.5%. This lower Fe removal efficiency is attributed to the presence of Fe mainly as Fe^{2+} in that AMD which required an additional oxidation step to convert Fe^{2+} to Fe^{3+} for a better removal of Fe.

Over 99% removal of Cu and above 90% removal of Zn are achieved by all four alkaline materials. The removal of Mn however was lower compared to the Cu and Zn removal. The lowest residual concentration of Cu after the alkalization is achieved with BFS ($<8.1 \times 10^{-7}$ mol/L) while the lowest residual concentration of Zn and Mn are achieved using NaOH (3.3×10^{-5} mol/L and 5.0×10^{-4} mol/L respectively) closely followed by BFS. Comparatively, better removal of Cu, Zn and Mn are achieved with BFS than with BA. This could be due to better sorption of those contaminants onto the surface of the BFS and on the precipitated mineral phases. With respect to the DAS PTS systems, a 92.6% Cu removal and around 80.4% Zn removal were achieved in the first limestone-DAS reactive tank at the Esperanza Mine treatment unit (Orden et al., 2021). The FAO limits for irrigation water of Cu, Zn and Mn should be 3.15×10^{-6} mol/L, 3.06×10^{-5} mol/L and 3.64×10^{-6} mol/L, respectively. For Cu, all alkaline materials except NaOH comply with this reference value. Yet, none of the used alkaline materials allowed to meet with residual Zn and residual Mn value indicated in FAO for irrigation water. However, it is noteworthy that the residual Zn concentration after NaOH and BFS alkalization are not far from the limit. The higher mobility of Mn in AMD treatments in comparison with other metals such as Cu and Zn has been previously reported (Orden et al., 2021). Additional treatment step(s) would therefore be required to improve such divalent metal removal. A multi-step alkaline PTS has been proposed with the use of MgO-DAS preceded by limestone-DAS to facilitate the increase of pH above 10 and thereby to achieve a better removal of divalent metals including Mn (Macias et al., 2012b).

Given the fact that above 99.9% of Fe and above 99.8% of Al are removed from the AMD, the formed solids should definitely contain Fe and Al, which is consistent with the SEM data. The precipitation of amorphous to poorly crystalline schwertmannite, i.e., a Fe containing oxyhydroxysulfate and basaluminite, an Al containing oxyhydroxysulfate are reported in the literature. (Sánchez-España et al., 2011; Protano FR, 2002; Lozano et al., 2020b; Park et al., 2016). This is also consistent with what is reported for the DAS PTS in the IPB (Orden et al., 2021). Geochemical modelling revealed that some forms of schwertmannite were still over-saturated in the supernatants after alkalization with NaOH and CaCO_3 (Table SI 4). This indicates those phases did not reach equilibrium yet in the supernatant and thus could be still in the process of precipitation. However, schwertmannite_2 after alkalization with NaOH and schwertmannite_4 after alkalization with CaCO_3 were closer to equilibrium (Table SI 4). None of the Fe phases were predicted after alkalization with BFS and BA as the respective Fe concentrations were below LoQ (Table SI 4). Similar to the schwertmannites, two forms of basaluminite were also over-saturated in the supernatants after alkalization regardless of the alkaline material used (Table SI 4) indicating those forms could also still be precipitating. Nevertheless, gibbsite (amorphous) phase was slightly oversaturated in all the supernatants indicating the possibility of Al hydroxides precipitation instead of Al oxyhydroxysulfates precipitation. The removal of the divalent metals, i.e., Cu and Zn, is likely due to co-precipitation/sorption with/on Fe and Al containing secondary mineral phases (Lozano et al., 2019b, 2020c; Caraballo et al., 2011). Cánovas et al. (2020) reported that Cu and Zn may be associated with basaluminite precipitation during AMD alkalization with limestone.

Comparatively, BA is more efficient for Fe and Al removal while BFS is more efficient for divalent metal removal as revealed by the residual concentrations in the supernatants after alkalization. This could be

due to a slightly higher pH at the end of BFS alkalization (pH 6.67) than that after BA alkalization (pH 6.32). Yet, overall both BFS and BA perform as good as, if not better than the currently used limestone and appear as good replacement options. However, the FAO for irrigation water criteria is not all met with these materials. Thus, additive treatment units like for example BaCO_3 -DAS and MgO-DAS should be considered (Gomez-Arias et al., 2015; Torres et al., 2018; Macias et al., 2012b).

3.3.2. REE removal after alkalization

The calculated REE speciation in the supernatants after alkalization is shown in Figure SI 2. It revealed that around pH 6.5, REE carbonated complexes (LnCO_3^+) could also be present in addition to the LnSO_4^+ , $\text{Ln}(\text{SO}_4)_2$ and Ln^{3+} . The percentage of LnCO_3^+ is the largest in the supernatant after alkalization with CaCO_3 among others while it is the lowest in the supernatant after alkalization with NaOH.

Fig. 5 presents the initial and residual concentrations of the REE.

Overall, REE removal is effective with all four alkaline materials used (>99.7%). These REE removal efficiencies are higher than what has been obtained in the field after the first limestone-DAS pond (i.e. 97.3%) at the Mina Esperanza mine (Orden et al., 2021). Contrary to transition metals, there are yet no REE limits. With respect to residual REE concentrations, alkalization with CaCO_3 yields lowest residual REE concentrations while the highest residual REE concentrations are obtained with NaOH alkalization. On average across all REE, residual REE are factor 2.5 higher in NaOH treatment than in the mean of BA, BFS and CaCO_3 . The removal of LREE is less efficient compared to MREE and HREE with all four alkaline materials. The residual LREE concentrations follow the order of $\text{CaCO}_3 < \text{BFS} < \text{BA} < \text{NaOH}$, whereas residual MREE concentrations follow the order of $\text{BFS} < \text{BA} < \text{CaCO}_3 < \text{NaOH}$. Meanwhile the residual HREE concentrations follow the order of $\text{BA} < \text{BFS} < \text{CaCO}_3 < \text{NaOH}$. Previous studies have suggested that REE removal during AMD alkalization in limestone-DAS PTS is mainly due to co-precipitation with Al containing secondary mineral phases (i.e., basaluminite) (Orden et al., 2021). Moreover, other studies showed also sorption of REE on Fe and Al mineral phases (above pH 5 on basaluminite (Lozano et al., 2019b) and above pH 6.5 on schwertmannite (Lozano et al., 2020a)). Gypsum may also be a host for scavenging REE which is suggested by the clearly lower removal efficiency obtained with NaOH, where gypsum was not formed, in comparison to the other alkaline materials. The ionic radius of Ca^{2+} in gypsum structure (in 8-fold coordination) is very similar to those of REE, there is indeed a possibility - especially for LREE - to fit into the gypsum structure as LREE^{3+} (Table SI 5). Such incorporation process was indeed identified by several researchers (Dutrizac, 2017; Lin et al., 2019; Sadri et al., 2018, 2019). It would therefore be worth investigating further the role of gypsum in REE scavenging during AMD alkalization.

In reality, PTS work on longer time scales. Therefore, ageing might also have an influence on the long-term retention of REE, i.e. incorporation or release from the structure with crystallization (Sánchez-España et al., 2011). Further studies using advanced characterization techniques (i.e., synchrotron-based techniques) and ageing experiments are needed for the confirmation of the retention mechanisms of REE.

3.4. Environmental and economic implications

Based on our results, the two tested waste materials are shown to perform as efficiently in remediating AMD and in retaining REE as commercial reagents. However, the amount of solid needed for an efficient neutralization of the AMD when using such waste materials, is higher than with CaCO_3 and NaOH. This is especially true for BA, which required in the current experiments eight times the amount of CaCO_3 and five times the amount of BFS to reach pH 6.5. One could argue that the need for a higher amount of solid would be a drawback in terms of transport and logistic perspective. However, with respect to environmental aspects, "recycling" BFS/BA in PTS support industrial symbiosis

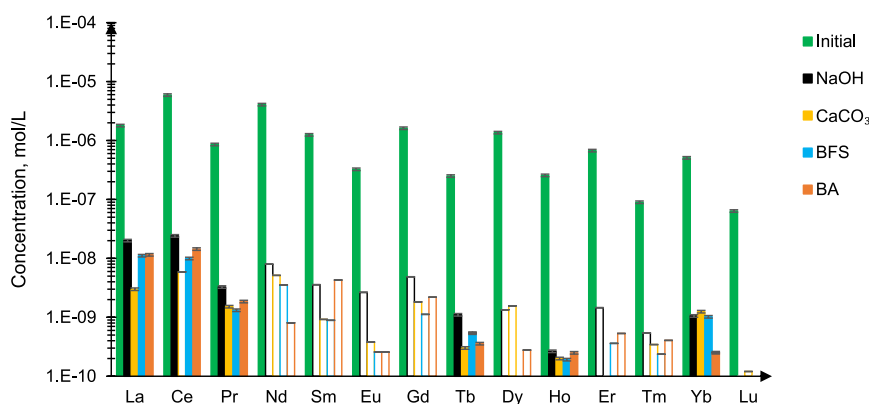


Fig. 5. The initial concentrations of REE and the residual concentrations of REE after the alkalization experiments (Unfilled bars: detection limit. Measured concentrations were below that limit. Means and standard deviation ($n = 2$ after treatment and $n = 3$ before treatment)).

and also circular economy. In fact, such associated transport and logistic costs could offset the costs associated with using commercial reagents and disposing of BFS/BA otherwise considered as waste. The DAS PTS at the Esperanza Mine alone used close to 25 tons of calcite per year which is close to around 1 million Euro per year (Orden et al., 2021; Millán-Becerro et al., 2021). The entire Tharsis mining area in the IPB contains approximately 5.2 hm³ of acidic water (Moreno González et al., 2019). Assuming AMD composition to be the same around the entire area, use of calcite as alkaline material is associated with more than 11 million Euro (considering the market price of 90–100 euro per ton).

Moreover, the possibility to retrieve critical elements of economic interest, such as REE from the solid precipitates, which still remains to be evaluated, would further help in offsetting the associated transport and logistic costs in using such waste materials improving the sustainability of such AMD PTS. This option has been proved suitable (Macías et al., 2017) using dilute commercial acids.

4. Conclusions

This study focused on comparing two different alkaline waste materials, i.e. BFS and BA with the classical alkaline agents CaCO₃ and NaOH for the remediation of highly acidic and metal rich AMD sampled in the IPB.

Based on the ANC determination and on the quantity of material needed to neutralize the pH of the AMD samples, it was found that BFS has a higher neutralizing capacity than BA, thus requiring less BFS addition than BA to reach a new neutral pH (pH = 6.5). Nevertheless, higher amounts of both materials are required for alkalization as compared to the classical reagents CaCO₃ and NaOH.

Overall, the two tested waste materials showed good potential for the remediation of AMD from the IPB. Both materials were efficiently able to neutralize the AMD and to remove Fe, Al and heavy metals. The Al and Fe comprised in the waste materials did not impend the remediation and the concentration of these two elements. They even decreased in the remediated solutions as compared to their concentrations after alkalization with CaCO₃ and NaOH.

With respect to their efficiency in removing REE, all four alkaline materials showed above 99.7% removal. However, with respect to the residual REE concentrations, the lowest were achieved using CaCO₃ for alkalization. Both BFS and BA can be considered as efficient REE remediating agents with a better performance of BFS. Indirect evidence suggest that gypsum may be a host for scavenging REE. The requirement for higher amounts of BFS and BA than using CaCO₃ for efficient remediation of the AMD could be viewed as beneficial in terms of environmental point of view. The associated transport and logistic costs of using such waste materials could be offset by 1) associated economic costs of using commercial materials such as calcite and 2) associated

economic benefits from being able to retrieving critical elements of economic interest such as REE. Further studies however are required for a correct estimation and confirmation of the latter. All those combined would improve the sustainability of AMD PTS leading to industrial symbiosis and thus circular economy.

The overall poorer performance of NaOH in remediating AMD in comparison to the other tested materials, highlighted that the efficiency of remediation is not only a matter of alkalization but also of the type of the precipitated phases. The presence of Ca in the alkaline material allowing gypsum precipitation was clearly shown to enhance the removal of heavy metals and REE. Yet, a lot of open questions remain to be answered on the exact nature of removal mechanism(s) along the alkalization process occurring inside PTS, especially for REE. Many removal mechanisms, such as co-precipitation, sorption (i.e. surface complexation) - alone or combined - could be responsible for the REE scavenge. Unravelling the fate of REE along the passive treatment system would therefore pave a path to enhance the concentration of REE in the precipitated solids for subsequent recovery at a later stage.

Declaration of competing interest

The authors declare that they have no known competing financial interests or personal relationships that could have appeared to influence the work reported in this paper.

Data availability

Data will be made available on request.

Acknowledgments

This research has received funding from European Union's Horizon 2020 research and innovation program under the Marie Skłodowska-Curie Grant Agreement No 857989. C.R Cánovas thanks the Spanish Ministry of Science and Innovation for the Postdoctoral Fellowship granted under application reference RYC2019-027949-I.

The authors gratefully acknowledge the valuable assistance of the following people as well: Dr. Raul Moreno Gonzalez from the Department of Earth Sciences, University of Huelva in Spain for assistance in collecting acid mine water samples; Dr. Quoc Tri Phung from SCK CEN in Belgium for assistance in obtaining BFS samples; Dr. Lander Frederickx from SCK CEN in Belgium for supporting in XRD analysis; Dr. Claudia Moens, Mr. Benoit Bergen and Ms. Kristin Coorevits, ICP-MS Team, Division of Soil and Water management, KU Leuven for their enormous assistance in measuring ICP-MS samples; All the technicians of the Waste and Disposal Group, SCKCEN for their assistance in various ways. Authors also thank ENCE Energía y Celulosa Company for

providing the BA.

Appendix A. Supplementary data

Supplementary data to this article can be found online at <https://doi.org/10.1016/j.apgeochem.2023.105800>.

List of abbreviations

AMD	Acid Mine Drainage
REE	Rare Earth Elements
BFS	Blast Furnace Slag
BA	Biomass Ashes
IPB	Iberian Pyrite Belt
ATS	Active Treatment Systems
PTS	Passive Treatment Systems
DAS	Dispersed Alkaline Substrate
μXRF	μ-X-ray fluorescence
FESEM-EDS	Field Emission Scanning Electron Microscopy with Energy-Dispersive X-ray Spectroscopy
EC	Electrical Conductivity
ORP	Oxidation Reduction Potential
ICP-MS	Inductively Coupled Plasma Mass Spectrometry
IC	Ion Chromatography
TIC/TOC	Total Inorganic Carbon/Total Organic Carbon
SEM/EDX	Scanning Electron Microscopy/Energy Dispersive Spectroscopy
XRD	X-ray Diffraction
ANC	Acid Neutralization Capacity
TD	ThermodemV1.10
TDB	Thermodynamic Database
LLNL	Lawrence Livermore National Laboratory
REY	Rare Earth Elements and Yttrium
EUS	European Shale
LREE	Light Rare Earth Elements
MREE	Middle Rare Earth Elements
HREE	Heavy Rare Earth Elements
FAO	Food and Agriculture Organization of the United Nations
LoQ	Limit of Quantification

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