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MINERALOGY AND PORE WATER CHEMISTRY OF A BOILER ASH

FROM A MSW FLUIDIZED-BED INCINERATOR

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Abstract:

This paper presents an investigation of the mineralogy and pore water chemistry of a boiler

ash sampled from a municipal solid waste fluidized-bed incinerator, subject to 18 months of

dynamic leaching in a large percolation column experiment. A particular focus is on the redox

behaviour of Cr(VI) in relation to metal aluminium Al⁰, as chromium may represent an

environmental or health hazard. The leaching behaviour and interaction between Cr(VI) and

Al⁰ are interpreted on the basis of mineralogical evolutions observed over the 18-month

period and of saturation indices calculated with the geochemical code PhreeqC and reviewed

thermodynamic data. Results of mineralogical analyses show in particular the alteration of

mineral phases during leaching (e.g. quartz and metal aluminium grains), while geochemical

calculations suggest equilibria of percolating fluids with respect to specific mineral phases

(e.g. monohydrocalcite and aluminium hydroxide). The combination of leaching data on a

large scale and mineralogical analyses document the coupled leaching behaviour of

aluminium and chromium, with chromium appearing in the pore fluids in its hexavalent and

mobile state once metal aluminium is no longer available for chromium reduction.

Key words: MSW ash, fluidized-bed, leaching, Al metal, Cr(VI), PhreeqC.

1. Introduction

Fluidized-bed combustion (FBC) is a well established technology for burning coal (Valk,

1994), sometimes in combination with biomass (Gogebakan and Selçuk, 2009) or for burning

relatively homogeneous wastes such as sewage sludge (Van de Velden et al., 2008, Toledo et

al., 2005). In recent years, the technology has sometimes been applied to the thermal

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treatment of municipal solid waste (MSW; Rhyner et al., 1995), but such an application remains an exception. In France for example, out of the 128 MSW thermal treatment units in operation since 2005, only 4 were based on the FBC technology, while the remaining units relied on the grate kiln (stationary or moving grate) technology.

The basic principle of FBC is the injection of air at the base of the combustion chamber through a layer of inert material (typically sand) which generates a suspended mass of waste and sand particles. This improves waste particle combustion due to the greater contact surface between the waste and the sand, with the drawback however that the waste must be shredded to a particle size of around 100 to 300 mm, following the type of technology used. This type of thermal treatment generates four solid residues: on the one hand air pollution control (APC) residues and bottom ash, typical of grate kiln incinerators, and on the other hand hopper ash and cyclone ash. Note that the ash studied in this paper and referred to as "boiler ash", is analogous to what Abbas et al. (2003) refer to as "hopper ash" (see Fig. 1 of Abbas et al., 2003). According to Abbas et al. (2003), among the advantages of the FBC technology are the low NOx and SOx emissions, the efficiency of heat transfer from the sand bed to the heat exchanger walls and the low amount of unburned matter in the ash residues.

As a result of the limited extent of FBC technology compared to grate kiln incinerators in the thermal treatment of MSW, relatively few studies have addressed the chemical composition and mineralogy of the residues. Abbas et al. (2003) examined the extraction, using batch leaching tests, of salts from filter ash, cyclone ash, hopper ash and bottom ash sampled from two FBC boilers in Sweden. Abbas et al. (2001) focused in particular on the leaching behaviour of chromium in different types of FBC ash (bottom, hopper, cyclone and filter ashes). These authors underline the influence of Al⁰ on chromium speciation and hence mobility.

The leaching behaviour of chromium in ash from waste thermal treatment plants has also been studied by Astrup et al. (2005; 2006). Astrup et al. (2005) provide a conceptual model for the interaction between Cr, Al and oxygen that was applied by Guyonnet et al. (2008) to reproduce the leaching behaviour observed during leaching experiments performed at several scales on a boiler ash from a FBC plant in France. This paper provides additional data for this same boiler ash with a special focus on the ash mineralogy and geochemical equilibria in pore waters collected during the leaching experiment performed in a large column. The new data provide additional insight into the composition and leaching behaviour of such residues.

2. Materials and Methods

2.1. Material preparation

The thermal process residue studied herein is a boiler ash from a French rotary fluidized-bed MSW incineration plant, which burns around 54,000 tons of waste annually. Eighteen tons of boiler ash were recovered in big bags and dried in a depot from an initial water content of around 10 weight percent to approximately 0.5%. Because preliminary studies of the ash's reactivity indicated a risk of setting during percolation experiments, it was mixed in equal weight proportions with washed Loire sand (0.1–1 mm). A horizontal continuous feed pugmill fed by two circuits; one for sand and one for ash, was used to guarantee homogeneous mixing between the two materials. The flowrates of the two circuits were checked approximately every 15 minutes. A flowsheet of the preparation of the ash-sand mixture can be found in Guyonnet et al (2005).

Following mixing, twenty measurements of the ash-sand dry density yielded an average value of $1.32~\text{g/cm}^3$ with a standard deviation of $0.035~\text{g/cm}^3$. However, during the filling up of the column experiment described below, some consolidation occurred and overall dry density increased slightly to $1.39~\text{g/cm}^3$. The grain size of the ash-sand mixture was below 2 mm, with a 10% fraction between $10~\text{and}~100\mu\text{m}$. Chemical and mineralogical analyses were performed on the raw ash and on the ash-sand mixture before and after 18~months of dynamic leaching.

2.2. Global chemical analyses of the solids

Chemical analyses were performed by ICP-AES after alkaline fusion using Na_2O_2 at 450°C and then digestion by HCl (method adapted from NF ISO 14869-2, 2002). Major elements (Ca, Si, Al, Fe, K, Mg, Mn, Na, P, Si, Ti) in the fresh ash were also determined by X-ray fluorescence (Philips PW2400, sequential) following sample preparation using the bead technique (addition of a fluxing agent; lithium tetraborate). Chloride was determined by potentiometric analysis. Total sulphur, total carbon and inorganic carbon were determined by non-dispersive infrared analysis of the gases CO_2 and SO_2 released following specific treatments: digestion by a 25% phosphoric acid solid for inorganic carbon, and oxidation under O_2 at 900°C for total carbon and total sulphur. The amount of organic carbon was derived by subtracting inorganic carbon from total carbon ($C_{org} = C_{tot} - C_{inorg}$).

Contents of Al⁰ and Cr(VI) in the solid were analyzed before and after leaching in the large column. Hexavalent chromium in the solid was analyzed colorimetrically following a reaction between Cr(VI) and 1,5-diphenylcarbohydrazide, under alkaline and warm conditions (EPA 3060A and 7196). Independent checks were performed by adding known amounts of K₂CrO₄ and Cr(III). Metal aluminium was measured by atomic adsorption spectroscopy following a controlled attack in a methanol-brome medium (Young, 1974). Crystallisation water was measured using the Penfield tube method (Sandell, 1951). Uncombined lime was determined by the classical ethylene glycol method (MacPherson and Forbrich, 1937).

2.3. Mineralogical characterisation of the fresh and leached ash

Crushed powders washed sand and five boiler ash grain-size fractions ($500\mu m - 2$ mm; 250-500 μm ; 125-250 μm ; 63-125 μm ; < 63 μm) were analysed by X-Ray Diffraction before leaching to identify major phases. The diffractometer with a Bragg Brentano geometry (Siemens D5000) is equipped with a monochromator and with a cobalt $K_{\alpha 1}$ radiation (λ = 1.78897 Å). Acquisition time for the XRD patterns in the 4-84°20 interval was set at one second per $0.02^{\circ}20$ step (=1 hour) for rotating samples. The phyllitic fraction of the sand (clays) was determined according to standard methodology on oriented samples with and without glycol during 12h and then heated at 490° C (4h). The crystalline phases were identified using Diffrac-AT software, in conjunction with the JCPDS (Joint Committee on Powder Diffraction Standards) database.

Micro-characterisation of the samples was performed in order to confirm XRD results with respect to major crystalline phases, to identify minor and amorphous phases and to detect trace elements in the phases by element mapping and quantitative analyses. Polished thin sections of fresh ash and leached ash-sand, were examined under a scanning electron microscope (SEM) with qualitative energy dispersive spectroscopy (EDS), and a Cameca SX 50 electronic microprobe (EMP) with quantitative wavelength dispersive spectroscopy (EWS). Additional raw samples of fresh ash were observed by SEM-EDS. EMP conditions for element mapping were an acceleration voltage of 15 kV and a reference current of 14 nA. EMP spot analyses were performed with an acceleration voltage of 15 kV and a reference current of 14 nA. Counting time was 6 seconds for the major elements and 20 seconds for the trace elements. The used standards were: albite for NaKα and SiKα, orthose for KKα, Al2O3 for AlKα, MnTiO3 for TiKα and MnKα, andradite for CaKα, Fe2O3 for FeKα, forsterite for

MgK α , vanadinite for ClK α , barytine for SL α , fluorapatite for PK α , Cr2O3 for CrK α , Cu metal for CuK α , PbS for PbM α , ZnO for ZnK α . Matrix effect corrections were performed with the "PAP" model (Pouchou and Pichoir, 1991).

2.4. Dynamic and static leaching tests –Water analyses

A downward-flow percolation experiment was performed in a large column described in Guyonnet et al. (2008). In summary the column is 5 metres high, 1 metre in diameter and equipped with an outlet at the bottom and 5 porous cups, noted PC1 through PC5, installed at 0.8, 1.82, 2.54, 3.55, and 4.36 m from the waste surface respectively. Data collected at different depths within the column provide an opportunity to investigate scale effects for a given infiltration rate. Infiltration of demineralised water was controlled automatically by sprinklers at the top of the column which sprayed 10 litres every 8 hours (30 litres/day) resulting, taking into account the column areas (one meter diameter), in an infiltration of approximately 3.8 cm/day. Porewater and effluent concentrations were monitored during 18 months, providing samples at final liquid/solid (L/S) values ranging from 3 L/kg at the outlet to 20 L/kg for the uppermost porous cup (PC1).

Prior to chemical analyses, all water samples were filtered at 0.45 μ m and sub-samples were acidified at pH 2 for cation analyses. Dissolved concentrations including total Cr were determined using standard methods by ICP-AES with a sequential Jobin Yvon JY 38 device (NF EN ISO 11885) and ionic chromatography (NF EN ISO 10304). Total inorganic carbon was analysed by infra-red spectroscopy on samples placed on a sealed tube with a septum in order to avoid carbonation. Cr(VI) was analysed by molecular absorption spectroscopy (NF T 90-043). Results showed that dissolved Cr was only present as Cr(VI) and therefore ICP-AES results on Cr_{total} were taken as Cr(VI) concentrations because of the lower quantification limit (5 μ g/l for the ICP-AES as compared to 10 μ g/l for the colorimetric method). Due to the relative abundance of aluminium, which may form colloids in waters at neutral values of pH, ultrafiltration at 4000 daltons was performed on several samples and analysed for Al, Ca, SiO₂ and SO₄. Results showed no significant differences with data measured in water samples filtered at 0.45 or 0.1 μ m.

At the end of the 18-month leaching period, the large column was carefully dismantled and samples were collected at different depths, along the central axis of the column, for analysis. Samples were collected and homogenised from each 1-metre section and analysed; a composite sample of the entire column was also created by quartering of these sub-samples.

The batch leaching test for granular waste at a liquid/solid (L/S) ratio of 10 L/kg was performed (CEN, 2002). During the dismantling procedure, it was observed that the material was loose and didn't show evidence of localized consolidation, concretions, colouring, etc., such as could be observed for example in bottom ash leaching experiments (Piantone et al., 2004). It is therefore hypothesized that, due to the material's homogeneity and grainsize, and to even distribution of flow at the column head, preferential flow was probably not significant.

3. Results and discussion

3.1. Global chemical composition before and after leaching

Table 1 completes data presented in Guyonnet et al. (2008) and provides a comparison with data from Abbas et al. (2001). Major elements in the ash sand mixture are related to the presence of Si-rich oxidized phases and to a lesser extent to the presence of Al, Ca, Fe and K. Trace elements in the ash-sand mixture are, by order of decreasing importance, Zn, Cu, Ba, Pb, Sn, Sr. Uncombined lime content (free and hydrated lime, respectively CaO and Ca(OH)₂) is low (< 0.5 mg/Kg). The proportion of zero-valent (metal) aluminium is 1.4-1.45 wt% (2 analyses), i.e., 23 wt% of the total aluminium. A similar characterisation of the waste after 18 months of leaching in the large column shows a decrease in the proportion of metal aluminium to 0.54 wt % suggesting that on average, 60% of metal aluminium has been altered. The significant increase of crystallisation water (0.7 to 1.7 % after 18 months of leaching) suggests the formation of hydrated minerals, probably of the ettringite type Ca₆Al₂(SO₄)₃.26H₂O as identified by SEM-EDS (Fig. 2b). As seen in Table 1, data from Abbas et al. (2001) measured on a FBC hopper ash, show values very similar to those measured in this study.

Analyses of Cr(VI) in the fresh ash (not included in Table 1) yielded values between 1.3 and 1.6 mg/Kg. But these values are most certainly underestimated since control tests with K₂CrO₄ and PbCrO₄ spikes showed that only 25% of the added Cr(VI) was recovered. Such analytical difficulties are also reported by Abbas et al. (2001) and are related to an analytical bias: the selective attack of the ash under alkaline and warm conditions, in presence of metal aluminium which is oxidized, promotes the reduction of Cr(VI) to Cr(III). Similar analyses on the ash-sand following 18 months of leaching yielded 3.5 mg/Kg Cr(VI). This result is believed to be reliable because independent checks with spikes of K₂CrO₄ and PbCrO₄

indicated total recovery of added Cr. This is explained by a "passivation" of metal aluminium Al^0 following the 18 months of dynamic leaching, which no longer promotes a reduction of Cr(VI) to Cr(III). This is confirmed by mineralogical analysis (next section) which shows a clear alteration of Al^0 grains and neoformation of Al/O precipitates following leaching, most probably $Al(OH)_3$ considering the alkaline pH. The clear redox relationship between aluminium and chromium is also clear from the porewater chemistry shown below.

3.2. Mineralogical phases

From XRD analyses of five grain-size fractions, the fresh ash contains primarily quartz, silicates (microcline, plagioclase, diopside, gehlenite), carbonates such as calcite, some sulphates (anhydrite), iron oxides (hematite) and halite. The background noise of the fresh ash diffractograms (Fig. 3), i.e., truly amorphous phases and crystalline phases present in proportions too small to be detected, is quite significant. Furthermore a diffuse bump centred around $34^{\circ}2\theta$ is clearly noticed for the limiting grain-size grain fractions (>500 μ m and <63 μ m) indicating the presence of poorly organized crystalline structures. This is confirmed by optical microscopy on the < 63 μ m fraction which showed numerous vitreous microspheres. From XRD analysis, the sand contains quartz (around 45%), microcline, plagioclase, augite and a small portion of clay of illite/mica type. The predominance of the phases was confirmed by SEM (Fig. 1), qualitative EDS analyses and quantitative EMP analyses.

Calcium phosphates were also identified by microscopic analyses (SEM-EDS and EMP). Note that XRD, which is typically used to detect crystalline phases above 1-5%, was not applied to samples collected after leaching in the large column because i) of the mixing with sand (50/50) and ii) the probably poorly crystalline state of neoformed minerals. Therefore direct microscopic analyses were preferred. The composition of calcium phosphates shows an increase of calcium content with leaching. While Ca/P ratios were initially close to 0.9-1.2, they increased to 1.4-1.6 at the end of the leaching experiment, confirming the greater stability of the resulting phosphates which probably evolved from monetite (Ca/P = 1) to the more stable whitlockite (Ca/P = 1.5) and finally apatite type (Ca/P = 1.87).

Micro-mappings of a large selection of elements present in the ash reveal that Al is mainly present as Al⁰ and Al hydroxyde, Ba as sulphate, Cu as Cu⁰, CuO or (Cu,Mn)O, Mn as (Mn,Zn)O, Ti as TiO₂ and Zn as (Zn,Mn)O. SEM images in Fig. 1 show that the quartz grains are coated with a Ca-bearing phase which may constitute the reactive part of the ash and is probably analogous to a cement-type phase. Microscopic analyses (SEM-EDS and EMP) of the waste after 18 months of leaching in the large column show the alteration of quartz and

feldspar grains (development of micro-fissures) as well as the precipitation of aluminium hydroxides and most probably of hydrated phases such as ettringite (Ca₆Al₂(SO₄)₃.26H₂O) (Fig. 2). The latter phase was largely suspected from SEM-EDS qualitative analyses (from Al, Ca and S contents) and from quantitative microprobe analysis performed on small grains which revealed a non-stoechiometric phase consistent with a mixture of phases such as Al(OH)₃, ettringite and calcium sulphates. Both ettringite and Al(OH)₃ are known to generate swelling in materials or roads during exothermic reactions. Although no significant swelling related to such phases was detected in the large column, swelling may have occurred as suggested by temperature variations measured in the large column during the first twenty days of leaching. Significant swelling was observed in lysimeter cells described in Guyonnet et al. (2008). Other cement-type phases, referred to below in the geochemical calculations, were not clearly identified.

3.3. Results of leaching tests

While data from the outlet of the large column were shown in Guyonnet et al. (2008), Fig. 3 shows the evolution over time of pH, Eh, Al and Cr(VI) measured in all the porous cups (PC) of the column, as well as at the outlet. The pH (11.5-12) is typical of alkaline incineration residues, while the redox potential (Eh) increased gradually from slightly reducing conditions (approximately -100 to 0 mV from the outlet to the uppermost porous cup) to more oxidising conditions at the end of the experiment (0 to 300 mV from the base to the top).

Figure 3 shows a remarkable inverse correlation between measured aluminium and chromium concentrations: Cr(VI) starts to appear when Al decreases. This was interpreted by Guyonnet et al. (2008), in agreement with Astrup et al. (2005), as a coupled effect between the oxidation of Al⁰ on the one hand and the reduction of Cr(VI) to Cr(III) on the other hand. It is only after approximately 130 days that Cr(VI) starts to appear in the uppermost porous cup (PC1). Before that time, chromium is in its reduced and less mobile state Cr(III). Once the metal aluminium has been "passivated" (Figs. 2c and 2d), aluminium concentrations decrease and chromium is mobilized as Cr(VI).

The porewater data collected from the large column is plotted in Fig. 4 as a function of depth at specific times. As seen in Fig. 4a, pH values remain high (> 11) throughout the column, but there appears to be a pH gradient with a slightly upward trend as depth increases. The redox potential measurements (Fig. 4b) suggests slightly more oxidizing conditions in the upper portion of the column (with an outlier at PC5; 4.36 m), as a result of the inflow of

oxygen-rich demineralised water fed at the top of the column. In Guyonnet et al. (2008) it was shown that it was necessary to account for the transport of dissolved oxygen (see Astrup et al., 2005) in order to reproduce the observed Al and Cr concentrations. The remarkable inverse correlation between Al and Cr(VI) is again illustrated by Figs. 4c and 4d.

Batch leaching tests performed at L/S=10 (CEN, 2002) on samples collected at different depths within the large column after 18 months leaching are shown in Table 2. The L/S values shown are the average values reached by the sampled sections over the 18-month period. Results obtained on the fresh ash-sand mixture are shown for comparison. Due to the liquid-solid ratios achieved in the column at the end of testing (more than 20 L/Kg at the top; 3 L/Kg at the bottom), the quantities leached from the fresh ash-sand mixture are generally much higher than from the samples collected from the large column. This does not appear to be the case however for Cr(VI), due on the one hand to the oxidation of Cr(III) to Cr(VI) during the percolation experiment, in accordance with the observed trend towards more oxidizing conditions (Fig. 3b), and on the other hand to an underestimation of the Cr(VI) concentration in the fresh ash-sand due to the presence of Al⁰.

In Table 2, the pH value of 10.3 measured during batch leaching of the fresh ash-sand mixture may seem low compared to the pH, around pH11, measured after leaching in the large column. However, this is consistent with the trend observed at the beginning of leaching in the large column (see Fig. 3a), where pH started quite low (around 8.7) and then increased with time and L/S ratio. It is known that the release of sodium and potassium, present in other minerals than salts, can occur with a delay in clinker/cement type materials (and depends largely on grain size). For example Taylor (1990) shows that, in the case of fresh cement pastes, this phenomenon gives higher pH values than in the case of equilibrium with portlandite Ca(OH)₂. Such a mechanism seems plausible for the high-temperature glass matrix of the ash and can help explain higher pH values at the end of leaching than in the fresh ash-sand mixture.

3.4. Geochemical equilibria

Thermodynamic calculations were performed in order to estimate saturation indices (SI) of the waters with respect to selected minerals, using version 2.12 of the PHREEQC geochemical code (Parkhurst and Appelo, 1999). Thermodynamic data are from the associated database Llnl.dat, with additional references for cement-type phases (Table 3) obtained from a review of thermodynamic data at different temperatures (Blanc et al., 2008).

Speciation calculations were performed on solutions with a charge balance error lower than 5%. Ionic strengths of waters were calculated. Considering the 0.5 eq/L limit given by Appelo et Postma (1999) for the Davies model activity used in PHREEQC outlet waters are to be considered only above L/S 0.3. All porous cup waters were validated (<0.5 eq/L).

Results in Fig. 5, plotted versus liquid-to-solid ratio (L/S), suggest that water samples collected at the outlet of the large column are in equilibrium with respect to monohydrocalcite, barite, wollastonite and aluminium hydroxide; Al(OH)₃ (Fig. 5a). Outlet waters are under-saturated with respect to gypsum and, as expected at a pH around 11.5, to portlandite. Saturation indices are also shown for some cement-type phases in Fig. 5b: values show important fluctuations for monocarbo-aluminate, calcium-aluminates (straëtlingite, hydrogarnet, C₄AH₁₃, Ca₂Al₂O₅:8H₂O) and ettringite, suggesting that these phases do not reach equilibrium within the bulk of the column during the experiment. However, precipitation of these phases could occur locally, considering heterogeneity in the large column, as indicated by Fig. 2b where ettringite was identified by microscopy.

Calculations performed with the chemistries measured in the samples collected from the porous cups (Fig. 6) suggest a trend towards equilibrium with respect to straëtlingite at high values of L/S value (>15), while waters become undersaturated with respect to Microcrystalline Gibbsite, Al(OH)₃ µC. In all cases, the samples appear to be in equilibrium with respect to monohydrocalcite. Hyks et al. (2009a and 2009b) also observed a trend towards equilibrium with respect to straëtlingite, in porewaters collected during column experiments with MSWI bottom ash and MSWI fly ash/APC residues.

The data presented herein confirm the conceptual model proposed by Astrup et al. (2005). Chromium appears at the outlet of the column once aluminium decreases in all the porous cups of the large column, suggesting the simultaneous oxidation of metal aluminium (Al^0) to Al(III) with a reduction of Cr(VI) to Cr(III):

$$Al^{0} + CrO_{4}^{2-} + 4 H_{2}O \rightarrow Al(OH)_{4}^{-} + Cr(OH)_{3}(s) + OH^{-}$$
 (1)

So long as there is Al⁰ to oxidize, chromium is in its non mobile trivalent form. However, as shown also by Astrup et al. (2005), two competitive reactions are also involved:

$$Al^0 + 0.75O_2 + 1.5 H_2O + OH^- \rightarrow Al(OH)_4^-$$
 (2)

$$Al^0 + OH^- + 3 H_2O \rightarrow Al(OH)_4^- + 1.5H_2$$
 (3)

Note that Eq. (3) involves the production of hydrogen. Hydrogen gas production was observed during batch ageing tests performed with the ash-sand mixture. Measured molar H₂ gas fractions were around 10% on average. Equations (2) and (3) tend to deplete the reducing capacity of Al⁰, leading to Cr mobility as Cr(VI). These mechanisms are consistent with observations in the large column where redox conditions evolved gradually towards more oxidizing conditions (Fig. 3b). As shown in Guyonnet et al. (2008), observed breakthroughs of chromium and aluminium in the column can be reproduced with a coupled chemistry-transport model (PHREEQC; Parkhurst and Appelo, 1999), taking into account Eqns. (1) through (3) and reaction kinetics.

4. Conclusions

The data presented herein illustrate the importance of mineralogical characterisation for a better understanding of the leaching behaviour of waste materials. Mineralogical and pore fluid equilibria calculations provide information regarding mineral phases controlling the solubilities of key elements such as Al and Cr, which can then be used for the quantitative interpretation of observed leaching behaviours, with the help of coupled chemistry-transport models (see Guyonnet et al., 2008). The mineralogical observations presented herein document the ageing process undergone by the ash during the leaching experiment, with an alteration of mineral phases such as metal aluminium or quartz and silicates. SEM images document the precipitation of hydrated neo-formed mineral phases such as ettringite or aluminium hydroxides, which can have negative implications for the reuse of the material (e.g. in roadworks), due to swelling.

The detailed pore water chemistry monitored during 18 months of leaching clearly illustrates the coupled behaviour of aluminium and chromium, with chromium appearing in percolating fluids in its mobile (hexavalent) form, when metal aluminium is no longer available for chromium reduction. Such breakthrough of chromium is an environmental and health concern and a potential reason why this boiler ash might not be accepted by environmental regulators as a candidate for utilisation (e.g. in roadworks). If it were not for this element, this ash could comply with the inert waste acceptance criteria (OJC, 2003), provided a "factor 3" is used for sulphate and chloride; OJC, 2003).

The data also have implications with respect to the interpretation of dynamic leaching experiments (e.g. CEN, 2005). Several authors (e.g. Grathwohl and Susset, 2009; Lopez-Meza et al., 2008; Hjelmar et al., 2007; Hjelmar, 1999) have shown that the liquid versus solid (L/S) ratio can be a very useful normalization parameter for comparing data from leaching tests performed on a given material at different scales and/or different percolation rates. This is not always the case however, as shown in Fig. 7 where Al and Cr data from Fig. 3 are plotted as a function of L/S ratio instead of time. As seen in this figure, the data do not collapse onto each other: due to the kinetics of the redox reactions that control the behaviour of these elements, leachate concentrations do not depend solely on the amount of water that has percolated through the waste per unit waste mass, but also on how much time metal aluminium has had to oxidize and on the stock of available reactive metal aluminium. This illustrates limitations of the L/S ratio as a normalization parameter in the case of reactive constituents controlled by reaction kinetics.

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Table 1
Global chemical compositions

| Wt% DM | Fresh sand | Fresh ash-sand* | Fresh ash** | Leached ash-sand* | Hopper ash*** | |
|--------------------------------------|------------|--------------------|-------------|-------------------|---------------|--|
| Al ₂ O ₃ total | 9.2 | 11.5 | 13.8 | 10.8 | 13.25 | |
| CaO | 0.9 | 7.6 | 14.3 | 8.7 | 7.68 | |
| Fe ₂ O ₃ total | 1.58 | 3.64 | 5.7 | 2.9 | 4.92 | |
| K_2O | 3.83 | 3.0 | 2.17 | 2.8 | 2.41 | |
| MgO | 0.6 | 1.3 | 2 | 1.4 | 1.30 | |
| MnO | 0.04 | 0.07 | 0.1 | 0.1 | 0.13 | |
| Na_2O | 1.7 | 1.7 | 1.7 | nm | 3.43 | |
| P_2O_5 | 0.1 | 1.25 | 2.4 | 1.1 | 1.07 | |
| SiO_2 | 80.8 | 67.4 | 54 | 60.4 | 64.35 | |
| TiO_2 | 0.32 | 0.69 | 1.06 | 0.6 | 0.93 | |
| Cl | n.m. | 0.43 | - | n.m. | n.m. | |
| C total | n.m. | 0.2 | - | n.m. | n.m. | |
| COT | n.m. | < 0.05 | - | n.m. | 0.08 | |
| S total | n.m. | 0.26 | - | n.m. | 0.24 | |
| LOI | | 2.16 | | | | |
| Al^0 | < 0.1 | 1.4 | 2.8 | 0.6 | n.m. | |
| H ₂ O crystal. | n.m. | 0.67 | - | 1.7 | n.m. | |
| mg/kg DM | | | | | | |
| Ag | < 0.2 | 4.3 | 8.6 | 4.3 | 0.8 | |
| As | 24 | 46 | 68 | 28 | 29 | |
| Ba | 680 | 955 | 1230 | 970 | 1300 | |
| Cd | < 2 | 4 | 8 | 3 | 6.4 | |
| Co | 7 | 61 | 115 | 59 | 19 | |
| Cr total | 28 | 168 | 308 | 185 | 283 | |
| Cu | 11 | 1021 | 2031 | 989 | 3150 | |
| Mo | < 5 | 8 | 16 | < 5 | 28 | |
| Ni | 28 | 63 | 98 | 55 | 88 | |
| Pb | 39 | 473 | 907 | 473 | 682 | |
| Sb | <10 | 44 | 88 | 52 | n.m. | |
| Sn | <10 | 155 | 310 | 234 | 123 | |
| Sr | 171 | 235 | 299 | 187 | 229 | |
| V | 18 | 24 | 30 | 33 | 44 | |
| Zn | 38 | 2053 | 4068 | 2268 | 2790 | |
| Zr | 64 | 48 | 32 | 91 | 177 | |

LOI = loss on ignition at 1000° C; DM = dry matter; n.m. = not measured; * = data from Guyonnet et al. (2007); ** = calculated as difference between ash-sand and sand; *** = data from Table 1 of Abbas et al. (2001)

Table 2 Results of batch leaching test (at $L/S = 10 \ L/Kg$), performed on samples collected at different depths from the large column after 18 months leaching

| Sampling depth (m) | Fresh* | 0 – 1 | 1 – 2 | 2-3 | 3 – 4 | 4 – 5 | Composite |
|---|--------|-------|-------|-------|-------|-------|-----------|
| Average L/S ratio after 18 months in column | - | 20 | 13.2 | 6.8 | 4.6 | 3.5 | 20 – 3.0 |
| рН | 10.3 | 10.7 | 10.7 | 10.85 | 10.8 | 10.85 | 11 |
| Conductivity (µS/cm) | 1580 | 303 | 300 | 365 | 368 | 378 | 340 |
| mg/Kg dry matter | | | | | | | |
| Al | 844 | 85.8 | 268.7 | 295.9 | 328.4 | 339.4 | 214.7 |
| Ca | 2312 | 669 | 700 | 811 | 891 | 909 | 1074 |
| Cl | 2697 | 14.1 | <10 | <10 | <10 | <10 | 18.7 |
| SO_4 | 1938 | 654 | 526 | 570 | 632 | 721 | 677 |
| Cr total | 0.27 | 0.41 | 0.21 | 0.25 | 0.3 | 0.32 | 0.47 |
| Cr(VI) | 0.26 | n.m. | n.m. | n.m. | n.m. | n.m. | 0.40 |

n.m.= not measured; * = data from Guyonnet et al. (2007)

Table 3

Thermodynamic constants of the main mineral phases considered in the geochemical modelling

| Phase | General formula and | log K | Ref.* | | | |
|---------------------------|---|-------------------------|--------|---|--|--|
| | cement nomenclature (in brackets) | | | | | |
| Carbonates | | | | | | |
| Aragonite | CaCO ₃ | | 1.99 | 1 | | |
| Barytocalcite | $BaCa(CO_3)_2$ | | 2.74 | 1 | | |
| Calcite | CaCO ₃ | [CC] | 1.85 | 1 | | |
| Strontianite | SrCO ₃ | | -0.31 | 1 | | |
| Monohydrocalcite | CaCO ₃ :H ₂ O | | 2.68 | 1 | | |
| Monocarbo-aluminate | Ca ₄ Al ₂ CO ₉ :10H ₂ O | [mCA] | 80.42 | 2 | | |
| Hemicarbo-aluminate | $Ca_8Al_4CO_{16}:21H_2O$ | [hCA] | 181.97 | 2 | | |
| Hydrated cement phases (C | CSH, CAH, CASH) | | | | | |
| Straëtlengite | Ca ₂ Al ₂ SiO ₇ :8H ₂ O | [C2A2SH8] | 49.71 | 3 | | |
| Hydrogarnet (katoite) | $Ca_3Al_2H_{12}O_{12}$ | [C3AH6] | 79.14 | 4 | | |
| C_4AH_{13} | $Ca_4Al_2O_{20}H_{26}$ | [C4AH13, AFm] | 102.02 | 5 | | |
| $Ca_2Al_2O_5:8H_2O$ | | | 59.57 | 1 | | |
| Sulphates | | | | | | |
| Anhydrite | $CaSO_4$ | | -4.30 | 1 | | |
| Barite | $BaSO_4$ | | -9.97 | 1 | | |
| Bassanite | CaSO ₄ :0.5H ₂ O | | -3.66 | 1 | | |
| Ettringite | $Ca_6Al_2(SO_4)_3(OH)_{12}:26$ | 6H ₂ O [AFt] | 57.08 | 6 | | |
| Gypsum | CaSO ₄ :2H ₂ O | | -4.48 | 1 | | |
| Other phases | | | | | | |
| Al(OH) ₃ μC | $Al(OH)_3$ | | 9.35 | 7 | | |
| Gibbsite | $Al(OH)_3$ | [AH3] | 7.76 | 1 | | |
| Portlandite | $Ca(OH)_2$ | [CH] | 22.81 | 8 | | |
| Quartz | SiO_2 | | -4.00 | 1 | | |
| Tridymite | SiO_2 | | -3.83 | 1 | | |
| Pseudowollastonite | CaSiO ₃ | [CS] | 14.00 | 1 | | |
| Wollastonite | CaSiO ₃ | [CS] | 13.76 | 1 | | |
| Larnite | Ca_2SiO_4 | [C2S] | 38.47 | 1 | | |
| Tenorite | CuO | | 7.66 | 1 | | |
| Zincite | ZnO | | 11.21 | 1 | | |
| Zn(OH) ₂ -beta | $Zn(OH)_2$ | 1.01 (1002-10) | 11.93 | 1 | | |

Ref 1: Llnl.dat; 2: Damidot et al. (1994); 3: Damidot and Glasser, (1993, 1995); 4: Damidot et al. (1992);

^{5:} Babushkin et al. (1985); 6: Perkins and Palmer, (1999); Möschner et al. (2008); 7: Eary, (1999); 8: Reardon, (1990)

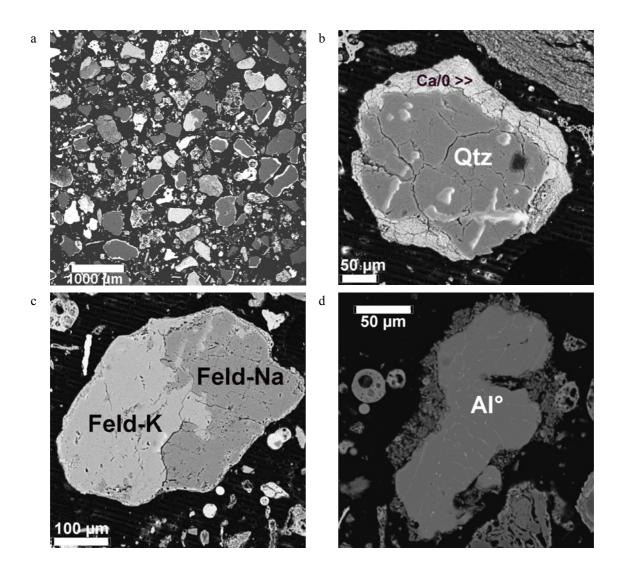


Fig.1. SEM images of fresh ash; (a) global ; (b) quartz with Ca-bearing coating ; (d) Feldspar ; (d) Al^0 .

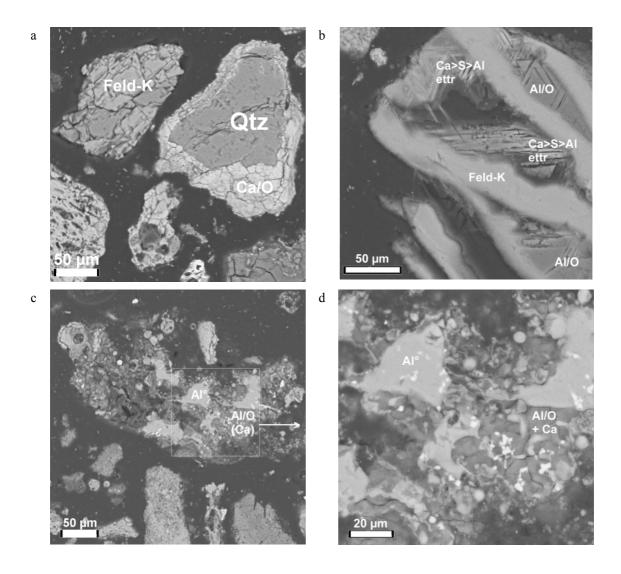


Fig.2. SEM images of ash-sand mixture after 18 months of dynamic leaching; (a) quartz and altered feldspar grains; (b) precipitation of Al(OH)₃ and ettringite (c) residual Al⁰ grain and (d) focus on precipitation of Al(OH)₃.

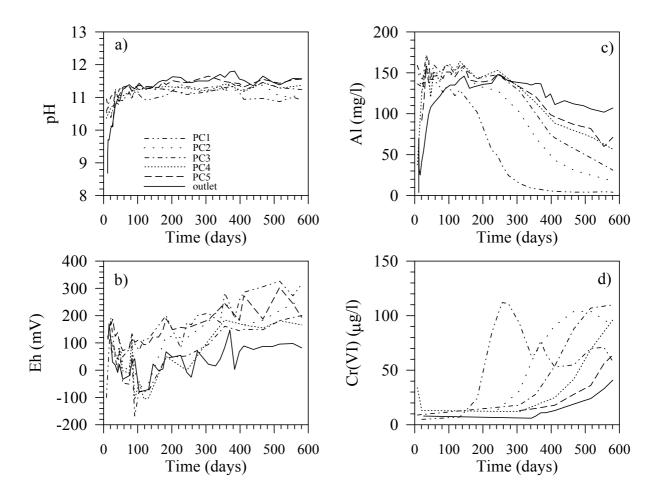


Fig.3. Evolution of (a) pH, (b) Eh, (c) Al and (d) Cr(VI) in the porewater at different depths in the column, as a function of time. PC1: 0.8 m; PC2: 1.82 m; PC3: 2.54 m; PC4: 3.55 m, PC5: 4.36 m; outlet: 5 m from column inlet.

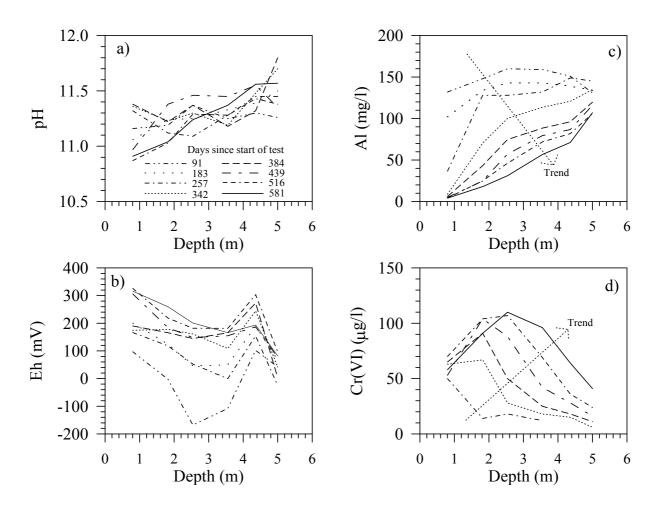


Fig.4. Evolution of (a) pH, (b) Eh, (c) Al and (d) Cr(VI) in the porewater at different times as a function of depth in the column.

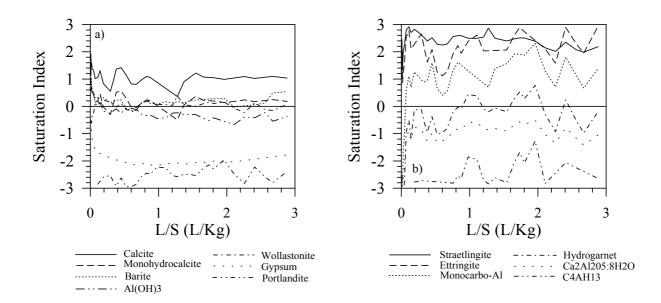


Fig.5. Evolution of saturation indices of 6 mineral phases, as a function of L/S ratio, calculated in samples at the outlet of the large column.

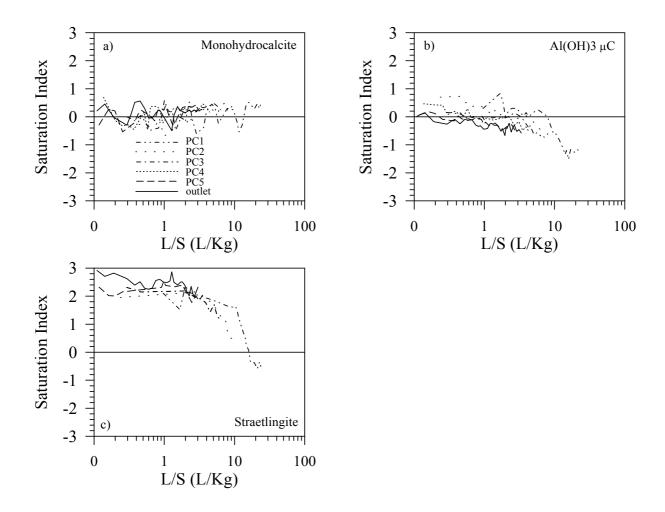


Fig.6. Evolution of saturation indices for three mineral phases (see Table 3); (a) Monohydrocalcite, (b) Micro-crystalline Gibbsite, (c) Straetlingite, as a function of L/S ratios, calculated in samples from the porous cups (PC) and at the outlet of the large column

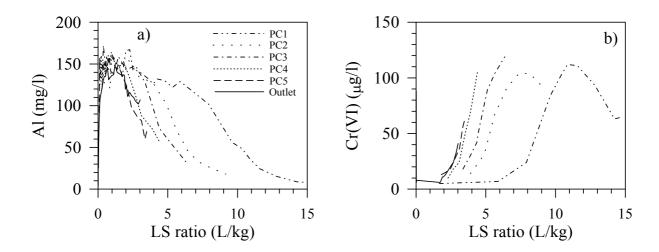


Fig.7. Evolution of (a) Al and (b) Cr(VI) in the porewater at different depths in the column, as a function of L/S ratio.