Cements based on kaolinite waste

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Abstract. The cement industry involves high-energy consumption that generates high CO₂ emissions into the atmosphere. Environmental concerns can be addressed by replacing parts of Portland cement clinkers with pozzolanic materials in mortars and concrete. Slag, fly ash and silica fume are materials considered for the planned replacement. Research studies on clay minerals, such as kaolinite, are being followed with special attention by the scientific community and the cement industry. It is well known that these minerals require an activation process to transform kaolinite (K) into metakaolinite (MK). MK is an amorphous material from the transformation of K with high pozzolanic activity, which is its capacity to react with the portlandite released during the hydration of Portland cement, generating compounds such as C–S–H gels and some aluminum-phase hydrates. One of the MK production methods is heat treatment controlled by kaolinite at temperatures in the range of 600–900 °C. Different residues have been used (coal mining, paper sludge and waste from a drinking water treatment plant) activated at 600 °C for 2 h to elaborate blended cements. Due to their good behaviour as future eco-efficient additions, this research is a study by x-ray fluorescence (XRF), x-ray diffraction (XRD) and scanning electron microscopy (SEM) of their influence on the performances of blended cement mixtures (binary and ternary one), with substitutions of pozzolan ratio at 28 days of hydration. The porosity of pozzolanic cements decreases because of the formation of hydrated phases during pozzolanic reaction.

1 Introduction

The cement industry traditionally uses in its production process a huge quantity of waste and by-products in different stages: alternative fuel, activated drinking water treatment waste, raw materials and/or pozzolanic additions in order to reduce the environmental impact. The use of cementitious materials in the manufacture of eco-efficient cements is focused on industrial waste. Research studies on clay minerals, such as kaolinite (K), are being followed with special attention by the scientific community and the cement industry (Rios et al., 2009; Vejmelkova et al., 2010; Janotka et al., 2010; Ptacek et al., 2014; Pesce et al., 2014; Frías et al., 2014; García-Giménez et al., 2016). It is well known that these minerals require an activation process to transform K into metakaolinite (MK), which is a highly pozzolanic product included in the European Community legislation (UNE-EN 197-1, 2011).

A way to obtain MK is through the activation of K-based industrial waste. Vigil et al. (2007) and Frías et al. (2008) reported that MK obtained from paper sludge, calcinated at temperatures between 650 and 700 °C for 2 h, has similar pozzolanic activity as other highly pozzolanic additions, such as commercial MK or silica fume, improving the performances of blended cements (Bai et al., 2003; Rodriguez et al., 2009). Another interesting research line to obtain recycled MK is the recycling of coal mining waste or activated drinking waste (Frías et al., 2014). These residues come from...
the processes of extracting and washing coal, and they are mainly composed of carbon and clay minerals (kaolinite).

The sludge generated in drinking water treatment plants is among this class of waste in significant quantities, which are normally taken to landfills for disposal with the ensuing environmental problems (Kynci, 2008).

The reuse of this drinking water sludge as a pozzolanic addition in the manufacture of new eco-efficient blended cements is a promising future alternative due to its silico-aluminous properties. The few works of reference in this area are by Gonçalves et al. (2004), who used blended cement mortars with substitutions of 5%, 10% and 20% of total cement content for a (kaolinite-based) drinking water treatment sludge from Portugal, and by Husillos et al. (2010), who characterised the behaviour of blended cement mortars with 10%–30% of atomised sludge (without kaolinite) of Spanish origin.

Blended cements properties depend on the nature and composition (CaO / SiO₂ ratio) of activated pozzolans. The research carried out in this paper included the study of scientific aspects (pozzolanicity, mineralogical phases), physical (porosity) and mechanical properties (compressive strength), and the relationship between them.

Due to their good behaviour as future eco-efficient additions, this paper conducts a detailed study of their influence on the performances of blended cement mixtures (binary and ternary), with substitutions of 10% and 20% pozzolan at 28 days of hydration.

2 Materials and experimental procedure

2.1 Materials

The materials used in this work are the following.

- Coal mining waste activated at 600°C during 2 h (ACW).

  The coal mining waste was a coal gangue waste (CW) deposited in a landfill at an opencast mine and collected by a Spanish coal group (Sociedad Anónima Hullera Vasco-Leonesa) in the Leon province, Spain. The coal mining waste was activated at 600°C for 2 h of retention in an electric laboratory furnace (the new samples are called ACW from now on), as this is the best activation temperature from an economic and energy point of view (Frias, 2012). ACW was ground in an agate mortar and pestle to obtain particle sizes less than 63 µm.

- Activated paper sludge (600°C during 2 h) + fly ash (APS + FA).

  APS was obtained by controlled calcination of paper sludge in our laboratory by means of an electric furnace at 600°C for 2 h, as described in previous work (García et al., 2010). The paper sludge (PW) was provided by Holmen Paper Madrid. This treatment eliminates organic material, achieves the total dehydroxylation of K into MK and avoids an excessive decarbonation of calcite, which could result in a significant amount of quicklime, a compound well known for its expansive property during hydration. APS and FA pozzolans were previously mixed with a ratio of 1:1 by weight, and the chemical composition of the mix is shown in Table 1. The FA met the requirements of American Society of Testing Materials (ASTM) Class F and the EN-UNE 450 specifications (UNE-EN 450-1, 2013): SiO₂ + Al₂O₃ + Fe₂O₃ contents higher than 70% and low CaO content.

- The starting waste (SW) for this research came from a Venezuelan drinking water treatment plant (Embalse La Mariposa). This wastewater treatment plant was initially around 92% RH. After drying, the ASW was activated at 600°C for 2 h of retention in an electric laboratory furnace at a heating rate of 20°C/min⁻¹, following the indications of a previous paper (Frias et al., 2012) in which these conditions were selected as the suitable activation conditions from an economic and energy point of view. The activated waste (ASW), when cooled to room temperature in a desiccator, showed a reddish coloration. The sample was ground in an agate mortar and pestle and then sieved through a 63 µm mesh.

- Ordinary Portland cement (OPC).

  The cement was a commercially available Ordinary Portland cement (OPC) classified as CEM I 52.5R supplied by Financiera y Minera (Italcementi group).

- Blended cements.

  Partial replacements of OPC with 10% and 20% pozzolan (ACW, APS + FA or AW) were made. The corresponding OPC–pozzolan mixtures were called 100 + 0, 90/10 and 80/20. Pastes were prepared at a demineralised water-to-binder ratio of 0.5. After mixing, a series of six prisms 1 × 1 × 6 cm were moulded and compacted by vibration. After 1 day at > 95% RH the samples were demoulded and cured at 21°C under > 95% RH in sealed containers for a period of 28 days from mixing.

2.2 Experimental procedure

The chemical composition of the samples was determined by X-ray fluorescence (XRF) using a Philips PW1480 dispersive energy spectrophotometer, Sc/Mo anode, in working conditions of 80 kV and 35 mA.

Mineralogical analyses were carried out by X-ray diffraction (XRD) using a Siemens D-5000 analytical powder diffractometer with an automatic divergence slit, graphite monochromator and Cu–Kα radiation. The XRD data were
collected in the angular range $3^\circ \leq 2\theta \leq 65^\circ$ with step 0.03$^\circ$ and counting time 3 s. The current and voltage intensities applied to the generating X-ray tube were 30 mA and 40 kW with divergence and receipt splits of 1 and 0.18$^\circ$, respectively. The complete identification and quantification were determined by the Rietveld method (De la Torre et al., 2001) with divergence and receipt splits of 1 and 0.18$^\circ$ with step 0.03$^\circ$ for $2\theta$ between $3^\circ$ and $65^\circ$. The results were obtained through chemical analyses using rutile as a standard sample.

The textural study of the material was accomplished using an electronic microscopy sweep with a Philips XL 30 and a Wolfram source. The powder samples were fixed to the metallic slide through a graphite plate (BIO-RED SC-502). The results were obtained through chemical analyses by EDX with a silicon–lithium detector and analyser DX41. Data shown in the tables are the average value of 10 measurements carried out in different fields.

3 Results and discussion

3.1 Characterisation of waste and activated waste

1. Coal mining

Quartz, calcite and phyllosilicates (mica illite type and kaolinite) compose the coal mining starting waste (CW). In addition, the coal mining waste activated at 600 $^\circ$C during 2h (ACW) is mainly by low quartz, micas and calcite (kaolinite peaks disappear after calcination) (Table 1). The reaction products after 28 days are layered double hydroxides (LDHs; formed from the tetrahedral and octahedral layers), stratlingite and the initial formation of C$_4$AH$_{13}$. The chemical composition is in Table 2.

2. Paper sludge waste

The mineralogical composition of paper sludge waste (PW) is calcite and phyllosilicates (talc and kaolinite). FA is formed by quartz, mullite and hematite. In the activated paper sludge (600 $^\circ$C for 2h) (50APS + 50FA) the main crystalline compounds are talc, calcite, mullite, quartz and hematite (Table 3); the peaks for the K$_2$O are no longer present after calcination. In the reaction after 28 days LDHs are shown. The chemical composition is given in Table 2.

3. Ordinary Portland cement (OPC)

The main crystalline compounds of the OPC are alite (42 % C$_3$S), belite (28 % C$_2$S), 21 % C$_3$A and brownmillerite (9 % C$_4$AF); the chemical composition appears in Table 2.

4. SW

SW was mainly composed of SiO$_2$ (36 %), Al$_2$O$_3$ (29 %) and Fe$_2$O$_3$ (10 %), followed by K$_2$O, MgO and TiO$_2$ with contents below 4 %, and the rest of the oxides were below 1 % (Table 2). The SW sample presented a total organic carbon content of 2.3 %, which was due to the presence of colloidal or dissolved organic matter in water. The mineralogical composition obtained by XRD (Table 4) shows the presence of phyllosilicates 2:1 (muscovite), phyllosilicates 1:1 (kaolinite), quartz, calcite, feldspars and hematite. In the reaction products after 28 days C$_4$AH$_{13}$ did not appear and of the LDHs only stratlingite was present.

The Frattini method, described in European Standard UNE-EN 196-5, was employed to determine the pozzolanicity activity of the blended cements. The test involves hydrothermal curing of the blended cements for 8 and 15 days at 40 $^\circ$C and 100 % RH. At the end of the curing period, the suspensions were analysed for Ca content and alkalinity. Each mixture was tested twice and the mean values are reported. Comparisons are made using the solubility curve of Ca(OH)$_2$ (UNE-EN 196-5, 2011).

<table>
<thead>
<tr>
<th>Minerals (%)</th>
<th>CW</th>
<th>ACW</th>
<th>ACW–lime 28 days</th>
</tr>
</thead>
<tbody>
<tr>
<td>Mica</td>
<td>20</td>
<td>14</td>
<td>n.d.</td>
</tr>
<tr>
<td>Kaolinite</td>
<td>14</td>
<td>n.d.</td>
<td>n.d.</td>
</tr>
<tr>
<td>Quartz</td>
<td>35</td>
<td>35</td>
<td>35</td>
</tr>
<tr>
<td>Calcite</td>
<td>17</td>
<td>13</td>
<td>16</td>
</tr>
<tr>
<td>LDH</td>
<td>n.d.</td>
<td>n.d.</td>
<td>6</td>
</tr>
<tr>
<td>C$<em>4$AH$</em>{13}$</td>
<td>n.d.</td>
<td>n.d.</td>
<td>3</td>
</tr>
<tr>
<td>Stratlingite</td>
<td>n.d.</td>
<td>n.d.</td>
<td>14</td>
</tr>
<tr>
<td>Amorphous material</td>
<td>14</td>
<td>38</td>
<td>26</td>
</tr>
<tr>
<td>$R_B$</td>
<td>10.20</td>
<td>9.57</td>
<td>8.32</td>
</tr>
<tr>
<td>$X^2$</td>
<td>5.80</td>
<td>4.79</td>
<td>4.47</td>
</tr>
</tbody>
</table>

The notation “n.d.” indicates not detected. LOI: loss of ignition.

<table>
<thead>
<tr>
<th>Oxide (%)</th>
<th>APS</th>
<th>FA</th>
<th>APS–FA</th>
<th>CW</th>
<th>ACW</th>
<th>SW</th>
<th>ASW</th>
<th>OPC</th>
</tr>
</thead>
<tbody>
<tr>
<td>SiO$_2$</td>
<td>13.9</td>
<td>55.7</td>
<td>34.8</td>
<td>49.8</td>
<td>56.6</td>
<td>36.2</td>
<td>41.5</td>
<td>20.2</td>
</tr>
<tr>
<td>Al$_2$O$_3$</td>
<td>8.3</td>
<td>24.0</td>
<td>16.2</td>
<td>21.7</td>
<td>25.3</td>
<td>29.5</td>
<td>33.6</td>
<td>4.3</td>
</tr>
<tr>
<td>Fe$_2$O$_3$</td>
<td>0.5</td>
<td>4.8</td>
<td>2.7</td>
<td>4.1</td>
<td>4.6</td>
<td>10.0</td>
<td>11.1</td>
<td>2.5</td>
</tr>
<tr>
<td>CaO</td>
<td>47.1</td>
<td>2.2</td>
<td>24.7</td>
<td>3.8</td>
<td>4.2</td>
<td>1.0</td>
<td>0.4</td>
<td>63.4</td>
</tr>
<tr>
<td>MgO</td>
<td>1.6</td>
<td>0.9</td>
<td>1.3</td>
<td>0.6</td>
<td>0.8</td>
<td>1.2</td>
<td>1.5</td>
<td>2.2</td>
</tr>
<tr>
<td>SO$_3$</td>
<td>n.d.</td>
<td>0.9</td>
<td>0.4</td>
<td>0.2</td>
<td>0.3</td>
<td>0.3</td>
<td>0.3</td>
<td>3.5</td>
</tr>
<tr>
<td>K$_2$O</td>
<td>0.3</td>
<td>2.2</td>
<td>1.2</td>
<td>2.7</td>
<td>3.1</td>
<td>3.3</td>
<td>3.7</td>
<td>0.9</td>
</tr>
<tr>
<td>Na$_2$O</td>
<td>0.2</td>
<td>0.5</td>
<td>0.3</td>
<td>0.1</td>
<td>0.2</td>
<td>0.8</td>
<td>1.0</td>
<td>0.3</td>
</tr>
<tr>
<td>TiO$_2$</td>
<td>0.3</td>
<td>0.7</td>
<td>0.5</td>
<td>1.1</td>
<td>1.2</td>
<td>1.2</td>
<td>1.4</td>
<td>0.2</td>
</tr>
<tr>
<td>P$_2$O$_5$</td>
<td>0.2</td>
<td>0.3</td>
<td>0.2</td>
<td>0.1</td>
<td>0.7</td>
<td>0.7</td>
<td>0.7</td>
<td>0.1</td>
</tr>
<tr>
<td>MnO</td>
<td>n.d.</td>
<td>n.d.</td>
<td>n.d.</td>
<td>0.1</td>
<td>0.1</td>
<td>0.4</td>
<td>0.4</td>
<td>n.d.</td>
</tr>
<tr>
<td>LOI</td>
<td>26.7</td>
<td>7.6</td>
<td>17.1</td>
<td>15.1</td>
<td>3.1</td>
<td>15.3</td>
<td>3.3</td>
<td>2.0</td>
</tr>
</tbody>
</table>

The notation “n.d.” indicates not detected. LOI: loss of ignition.
Table 3. Mineralogical composition by XRD of products from coal mining (PW, APS, FA, 50APS–50FA and after 28 days of reaction in the 50APS–50FA–lime system).

<table>
<thead>
<tr>
<th>Minerals (%)</th>
<th>PW</th>
<th>APS</th>
<th>FA</th>
<th>50APS–50FA</th>
<th>50APS–50FA–lime 28 days</th>
</tr>
</thead>
<tbody>
<tr>
<td>Talc</td>
<td>3</td>
<td>6</td>
<td>n.d.</td>
<td>3</td>
<td>n.d.</td>
</tr>
<tr>
<td>Quartz</td>
<td>n.d.</td>
<td>n.d.</td>
<td>31</td>
<td>15</td>
<td>15</td>
</tr>
<tr>
<td>Calcite</td>
<td>78</td>
<td>82</td>
<td>n.d.</td>
<td>41</td>
<td>20</td>
</tr>
<tr>
<td>Mullite</td>
<td>n.d.</td>
<td>n.d.</td>
<td>28</td>
<td>12</td>
<td>12</td>
</tr>
<tr>
<td>Hematite</td>
<td>n.d.</td>
<td>n.d.</td>
<td>26</td>
<td>12</td>
<td>6</td>
</tr>
<tr>
<td>LDH</td>
<td>n.d.</td>
<td>n.d.</td>
<td>n.d.</td>
<td>28</td>
<td>12</td>
</tr>
<tr>
<td>Amorphous material</td>
<td>11</td>
<td>12</td>
<td>15</td>
<td>17</td>
<td>31</td>
</tr>
<tr>
<td>$R_b$</td>
<td>10.20</td>
<td>9.57</td>
<td>10.49</td>
<td>11.36</td>
<td>7.52</td>
</tr>
<tr>
<td>$X_2$</td>
<td>5.80</td>
<td>4.79</td>
<td>5.24</td>
<td>6.21</td>
<td>4.99</td>
</tr>
</tbody>
</table>

The notation “n.d.” indicates not detected.

Figure 1. Evolution of the pozzolanic activity from the activated products.

3.2 Pozzolanicity of blended cements

The results obtained by the Frattini test for all blended cement pastes and for the reference cement paste at 15 days show that the only cement that could be considered as pozzolanic cement (Type CEM IV), according to the existing standard, is the cement paste elaborated with 20 % APS + FA, but not at 8 days (Fig. 1). However, this waste presents pozzolanic activity when analysed in a pure pozzolan–Ca(OH)$_2$ system (Rodríguez et al., 2009).

3.3 Mechanical properties

3.3.1 Compressive strength of blended cements

Figure 2 shows compressive strength at 2 and 28 days (average of six specimens) for all the studied blended cements and the reference cement. All the blended cements have a lower compressive strength than OPC at 2 days of hydration due to moderate pozzolanic activity. ACW blended cement pastes exhibited slightly higher compressive strength than the OPC paste at 28 days. In pastes containing APS + FA, the compressive strength is lower at this age for 10 % addition, probably due to the reactivity of FA being slower than MK-based pozzolans; however, the value for 20 % replacement is very close to that obtained for OPC paste.

3.3.2 Total porosity of blended cements

The total porosity of blended cements is directly related to the compression strength results of the pastes, as can be seen in Fig. 2; porosity decreases from 2 to 28 days of curing as compressive strength increases, which decreases as a consequence of the formation of hydrated phases during pozzolanic reaction. C–S–H gels appeared in all waste. APS + FA waste shows LDH compounds and C–S–H gels (Fig. 3a); the same compounds are present in the ACW (Fig. 3b) and stratlingite (Fig. 3c). The ASWs are micas, stratlingite and C–S–H gels (Fig. 3d). Due to the reduction of pores and because of the growth of hydrated compounds that have hydraulic properties, the compressive strength increases (Rodríguez et al., 2009). OPC samples are less porous than...
the samples with substitution due to moderate pozzolanic activity.

The chemical analysis by EDX of the phases at 28 days of reaction is presented in Table 5.

### Table 5. Chemical analysis by EDX of the phases at 28 days.

<table>
<thead>
<tr>
<th>Oxides (%)</th>
<th>C$_2$ASH$_8$</th>
<th>C–S–H gels</th>
<th>Mica</th>
<th>LDH</th>
<th>C$<em>4$AH$</em>{13}$</th>
</tr>
</thead>
<tbody>
<tr>
<td>Na$_2$O</td>
<td>n.d.</td>
<td>2.48 ± 0.07</td>
<td>2.37 ± 0.18</td>
<td>n.d.</td>
<td>n.d.</td>
</tr>
<tr>
<td>MgO</td>
<td>n.d.</td>
<td>1.91 ± 0.21</td>
<td>1.88 ± 0.22</td>
<td>n.d.</td>
<td>n.d.</td>
</tr>
<tr>
<td>Al$_2$O$_3$</td>
<td>34.04 ± 1.88</td>
<td>25.58 ± 0.43</td>
<td>32.32 ± 1.35</td>
<td>26.04 ± 0.92</td>
<td>34.48 ± 1.33</td>
</tr>
<tr>
<td>SiO$_2$</td>
<td>28.31 ± 1.63</td>
<td>34.09 ± 1.18</td>
<td>47.10 ± 2.05</td>
<td>40.39 ± 0.67</td>
<td>1.65 ± 0.06</td>
</tr>
<tr>
<td>K$_2$O</td>
<td>0.82 ± 0.76</td>
<td>2.02 ± 0.07</td>
<td>3.07 ± 0.54</td>
<td>n.d.</td>
<td>n.d.</td>
</tr>
<tr>
<td>CaO</td>
<td>29.76 ± 0.66</td>
<td>28.40 ± 1.59</td>
<td>6.06 ± 0.41</td>
<td>33.56 ± 0.89</td>
<td>63.87 ± 1.11</td>
</tr>
<tr>
<td>TiO$_2$</td>
<td>n.d.</td>
<td>n.d.</td>
<td>0.91 ± 0.11</td>
<td>n.d.</td>
<td>n.d.</td>
</tr>
<tr>
<td>Fe$_2$O$_3$</td>
<td>7.07 ± 0.47</td>
<td>5.50 ± 3.25</td>
<td>6.27 ± 0.82 n.d.</td>
<td>n.d.</td>
<td>n.d.</td>
</tr>
<tr>
<td>Total</td>
<td>100</td>
<td>100</td>
<td>100</td>
<td>100</td>
<td>100</td>
</tr>
</tbody>
</table>

The notation “n.d.” indicates not detected.

According to the Frattini test the only blended cement that can be considered as pozzolanic is the one elaborated with the addition of 20% ACW.

Hydrated calcium aluminate and LDH compounds are favoured with the addition of ACW. LDH compounds with the addition of APS + FA and stratlingite appeared in the ACW and ASW.

The value of compressive strength for blended cements with 20% replacement is very close to that obtained for OPC paste, although higher values were obtained with the addition of 20% ACW.
Due to the reduction of pores and because of the growth of hydrated compounds that have hydraulic properties, the compressive strength increases. Porosity decreases as a consequence of the formation of hydrated phases during pozzolanic reaction.

**Data availability.** No data sets were used in this article.

**Author contributions.** RVdlVM analysed the XRD and SEM data; RGG wrote the paper and analysed the XRD and SEM data; IVR performed the experiments; LFC analysed the FT-IR data; SMR analysed the TG and DTA data; MF designed the study.

**Competing interests.** The authors declare that they have no conflict of interest.

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