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Why comparison between different chemical extraction procedures is necessary to better assess the metals availability in sediments


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ABSTRACT

Single and sequential extractions are current and useful tools for estimating the availability of metals in soils or sediments. Many chemical extraction procedures have been proposed in the literature, making the comparison difficult. This study compares the data consistency of the potential availability given by four chemical extractions commonly found in the literature:

➢ two single procedures (using dilute HCl or EDTA as reactant)
➢ two sequential procedures (the modified standardized 3-steps procedure of the Standards, Measurements and Testing Programme (SMTP), namely BCR-extraction, and the 7-steps procedure of Leleyter and Probst (1999), namely LP-extraction).

The leaching procedures are all performed on the reference lake sediment CRM BCR-701, used as a reference to assess the accuracy of the modified standardized procedure of Rauret et al. (1999). The results show similar availabilities between HCl and EDTA single leaches for Zn, Cu, Cr and Pb and between HCl and the BCR procedures for Zn, Cu, Cr and Ni. The LP-extraction is the most aggressive of the four tested procedures (except for Pb extraction), mainly due to a better extraction of the exchangeable fraction, by a better dissolution of the acido-soluble phase and particularly to a better dissolution of the amorphous and crystalline Fe-oxides. The comparison between the four procedures reveals a major problem of Pb extraction, potentially due to the formation of lead precipitates during HCl and the LP-extractions, even if an overestimation of the lead availability by the BCR procedure cannot be excluded.

1. Introduction

In soils and sediments, metals and metalloids are distributed between different geochemical fractions, according to their nature, their chemical speciation and their reactivity. In order to assess their distribution between residual and non-residual fractions and their environmental availability i.e. their ability to be available to living organisms in case of changes of environmental parameters, many authors use chemical extractions (Da Silva et al., 2002; Giancoli Barreto et al., 2004; Gismera et al., 2004; El-Azim and El-Moselhy, 2005; Singh et al., 2005). Those procedures are based on the ion-exchange reactions, the dissolution of the compounds of the solid phase and the selectivity and specificity of the used chemical reagents. Depending on the purpose of the study, two kinds of extractions are possible: single or sequential extractions.

Single extractions are rapid, simple and cost-effective ways to investigate the labile metal fraction of soils and sediments. Reagents used generally fall into three categories: acids, un-buffered salts and complexing reagents (Leleyter et al., 2012; Leleyter, 2017). Dilute strong acids are often used to estimate the mobile fractions of soils or sediments (Agemian and Chau, 1976). They partially extract exchangeable elements and dissolved trace elements associated with carbonate, Fe/Mn-oxides and organic matter fractions (Rauret, 1998) and thus do not have a great selectivity. Among them, hydrochloric acid has been extensively used (Yu et al., 2021). It is assumed to extract metals due to
its acidic properties combined with the chelating properties of Cl\(^{-}\). The extraction efficiency of HCl is dependent on the concentration and extraction time (Snake et al., 2004; Yu et al., 2021). The HCl concentrations differ within separate studies: 0.1 M (Giancoli Barreto et al., 2004; Yu et al., 2004; Wei et al., 2005; Menzies et al., 2007; Hyeop-Jo Han et al., 2020; Fu et al., 2020; Gil-Diaz et al., 2021), 0.2 M (Kuo et al., 2006; Leleyter et al., 2012), 0.5 M (Agemian and Chau, 1976; Gaiero et al., 2003; Sutherland et al., 2004) or 1 M (Snake et al., 2004; Hamdoun et al., 2015b; Yu et al., 2021). EDTA is one of the most widely used complexing reagent because of its high extraction capacity (Sahuquillo et al., 2003). It is assumed to extract metals on exchange sites on both inorganic and organic complexes. However, it can also dissolve calcareous material and apatite minerals through the complexation of calcium and magnesium (Chao, 1984; Sahuquillo et al., 2003; Gismera et al., 2004). Most authors use the same EDTA concentration value (0.05 M) (McGready et al., 2005; Camizuli et al., 2014; Hamdoun et al., 2015b; Sow et al., 2018; Avunardi et al., 2019; Ben Abdelkader et al., 2019), even if 0.02 M EDTA is also reported (Gismera et al., 2004).

Sequential extractions procedures (SEP) are based on the use of several reagents applied to a solid sample aliquot. Successive specific leaching of geochemical fractions is performed, from the least aggressive to the most destructive reactants. Contrary to the single extractions, those procedures allow to characterize the different mobile fractions, even if they do not give any direct information about mineralogy. Various protocols were described in the literature (see review in Filgueiras et al., 2002 and Gleyzes et al., 2002). That is why in the framework of the Standards, Measurements and Testing Programme (SMTP), Quevauviller et al. (1994) then Rauret et al. (1999) proposed a harmonization of extraction protocols and achieved to a standardized 3-steps procedure (the BCR-extraction) which selectivity, reproducibility and repeatability were confirmed by further studies (Whalley and Grant, 1994; Davidson et al., 1998; Rauret et al., 1999). In this procedure, various certified materials (CRM 483, 601, 700 and 701) can be used in order to validate the extraction process (Quevauviller, 2002). This protocol has been widely adopted and applied to diverse media (see review in Sutherland, 2010; Kumkrong et al., 2020 and references therein), in particular since it is a compromise between one-step leaches and other multi-steps procedures. However, this extraction scheme could lead to an underestimation of metal mobility (Caraballo et al., 2018) and does not provide detailed information to scientists as some of the other sequential extraction procedures (Alan and Kara, 2019). In spite of the criticisms of their lack of selectivity and processes of re-dissolution of elements (Gomez-Ariza et al., 2000; Gleyzes et al., 2002), sequential extraction protocols, and more generally chemical extractions, are commonly used through the literature, mainly because there are relatively easy to implement for most of laboratories (see review in Filgueiras et al., 2002).

The sequential extractions are widely considered as effective analytical procedures to estimate the geochemical forms of metals and metalloids present in soils and to provide information on their geogenic or anthropogenic origin as well as on their availability towards other environmental compartments (Naz et al., 2020; khelifi et al., 2020). Naz et al. (2020) highlighted that the sequential extraction remains vital to assess and quantify the metal fractions/forms in the soil in order to understand the potential for metal uptake by plants, the bioavailable portion and its bioaccumulation. Gil-Diaz et al. (2021) reminded that most of the studies on risk assessment use metal(loid)s availability data from single or sequential extractions although the results may vary with the extractant solutions used. In their opinion, further studies comparing the metals availability estimated by various chemical procedures for a specific site are necessary. According Padoan et al. (2020), the use of chemical extractions provides useful indications to estimate the metals release to the environment in the case of flooding episodes.

Studies of selectivity must be applied to each new procedure or modification of existing procedures that is only weakly reported in the studies. To our knowledge, only Leleyter and Probst (1999) have developed an optimized sequential chemical extraction procedure, checked for efficiency, selectivity, reproducibility and repeatability of the different steps.

The objective of the present paper is to check the consistency of data on the mobility and the partitioning of Cr, Cu, Ni, Pb and Zn in the lake reference sediment (CMR code: BCR-701), given by four extraction procedures commonly found in the literature, i.e. two single extractions (dilute acid (HCl) and complexing agent (EDTA)) and two protocols of sequential extraction. The first one is the procedure from Leleyter and Probst (1999), frequently used on soils and sediments in various conditions (Leleyter and Probst, 1999; Aubert et al., 2004; Leleyter and Baraud, 2006; Cecchi et al., 2008; Salvareddy-Aranguen et al., 2008; Bur et al., 2009; N’Guessan et al., 2009; Leleyter et al., 2012; Seh-Bardan et al., 2012; El Azzi et al., 2013; Roussiez et al., 2013; Hamdoun et al., 2015a, 2015b; Baraud et al., 2017; Leleyter et al., 2018). The second one is the BCR procedure. Major elements such as Al, Fe, Mn and Ca are also investigated since those elements provide information on the dissolution of the major soil components, i.e. clays, oxides and carbonates.

2. Material and methods

2.1. Extraction procedures

All the reagents used to perform the extraction procedures or the mineralization of the total sample and the residual fraction are ultrapure grade and supplied by VWR®.

2.1.1. Single extraction

Two single extraction procedures are performed in triplicate on the BCR 701. The samples are shaken using a ratio 10:1 (liquid/solid), with either 0.5 mol·L\(^{-1}\) HCl (Kubová et al., 2008) or 0.05 mol·L\(^{-1}\) EDTA (Chao, 1984), during 1 h, at room temperature. Following centrifugation at 3000g during 15 min then filtration at 0.45 μm using PVDF hydrophilic membrane (HVLP filter), the solutions are stored at 4 °C until chemical analysis. The obtained labile fractions are named HCl-F\(_\text{lab}\), and EDTA-F\(_\text{lab}\), respectively.

2.1.2. Sequential extraction procedures

Two procedures are applied in triplicate. The BCR-extraction is the three steps BCR sequential procedure described by Rauret et al. (1999) and Pueyo et al. (2001) (see Table 1). Extracts are centrifuged at 3000g for 20 min and the supernant is decanted and transferred in polypropylene containers previously washed with 0.37 mol·L\(^{-1}\) HNO\(_3\). This procedure allows three operationally-defined non-residual fractions to be distinguished successively (BCR-F1 = acid-soluble, BCR-F2 = reducible, BCR-F3 = oxidizable) and one residual fraction (BCR-F4 = residual). The sum of BCR-F1 to F3 is the labile fraction, namely BCR-F\(_\text{lab}\).

The second procedure (LP-extraction) is the optimized sequential chemical extraction procedure developed by Leleyter and Probst (1999). This one allows seven non-residual mineralogical fractions to be distinguished successively (LP-F1a = water soluble, LP-F1b = exchangeable, LP-F1c = acid-soluble, LP-F2a = Mn-oxides, LP-F2b = amorphous Fe-oxides, LP-F2c = crystalline Fe-oxides, LP-F3 = oxidizable) and the residual fraction (LP-F4 = residual). Details on the reagents and on the physico-chemical parameters of each step are indicated in Table 1. After each step, extracts are filtrated through a 0.45 μm HVLP filter. Filtrates are transferred in polypropylene containers previously washed with 2% HNO\(_3\). The sum of the 7 non-residual fractions (LP F1 to F3) is the labile fraction, namely LP-F\(_\text{lab}\). Following each step of both procedures, the solutions were stored at 4 °C until chemical analysis.

On the residue as well on the bulk sample, a four acids digestion procedure is applied as described below and in Table 1. Total digestion is performed as the following steps:
Table 1
Sequential extraction performed according to the procedures of Rauret et al. (1999) and Leleyter and Probst (1999). RT: room temperature.

<table>
<thead>
<tr>
<th>Desired fraction</th>
<th>Extractive reagent</th>
<th>Reaction time and temperature</th>
<th>Reagent/sample ratio</th>
</tr>
</thead>
<tbody>
<tr>
<td>BCR</td>
<td>Rauret et al., 1999</td>
<td></td>
<td></td>
</tr>
<tr>
<td>LP</td>
<td>Leleyter and Probst, 1999</td>
<td></td>
<td></td>
</tr>
<tr>
<td>F1 — acid-soluble</td>
<td>F1a = water-soluble</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>F1b = exchangeable</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>F1c = acido-soluble</td>
<td></td>
<td></td>
</tr>
<tr>
<td>F2 — reducible (isop)</td>
<td>F2a = Mn-oxides</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>+ Fe-oxides</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>F2b = amorphous Fe-oxides</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>F2c = crystalline Fe-oxides</td>
<td></td>
<td></td>
</tr>
<tr>
<td>F3 — oxidizable</td>
<td>F3a = oxidizable</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>F3b = oxidizable</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>F3c = oxidizable</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>I. Evaporation of the solution at 70 °C</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>1 h, room temperature + 1 h, 85 °C</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>1 h, room temperature + 1 h, 80 °C</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>1 h, room temperature + 1 h, 70 °C</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>2 days, 70 °C</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>2 days, 120 °C</td>
<td></td>
<td></td>
</tr>
<tr>
<td>Residue and bulk sample</td>
<td>HF/HNO3/H2O/H2O2</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>HClO4</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>H2BO3 + HCl 6 N + Milli Q water</td>
<td></td>
<td></td>
</tr>
<tr>
<td></td>
<td>HNO3 7.5</td>
<td></td>
<td></td>
</tr>
</tbody>
</table>

1. Microwave digestion of 0.250 g of dry residue or bulk sample with a solution of HF/HNO3/HCl during 20 min at 200 °C
2. Evaporation of the mixture at 70 °C.
3. Dissolution and homogenization of the dry residue in 1 mL of HClO4 + H2O2 30% v/v, heating at 120 °C during 2 days.
4. After cooling, evaporation of the mixture in 3 drying steps: 8 h at 120 °C, 12 h at 165 °C and 1–2 h at 190 °C to avoid to form fluorides which could coprecipitate trace elements and to be insoluble (Yokoyama et al., 1999).
5. Dissolution and homogenization of the dry residue in 1 mL of HBrO4 + 2 mL of HCl 6 mol-L⁻¹ + 3 mL of MQ Water, heating at 110 °C during 1 day + 1 night.
6. Evaporation of the solution at 70 °C.
7. Dissolution and homogenization of the dry residue in 10 mL of HNO3 7.5 mol-L⁻¹, heating at 70 °C during 1 day.
8. Storage at 4 °C until chemical analysis.

2.2. Certified reference sediment

Single and sequential extractions are performed on the certified reference material BCR-701. This material has been collected from Lake Orta (Piemonte, Italy), which is known for serious metal contamination due to industrial discharges (Provinzi and Gaggino, 1986; Baudo et al., 1989). Details on the preparation and the certification of the BCR-701 sediment are found in Pueyo et al. (2001), Quevauviller (2002) and Kubová et al. (2004). Certified concentrations of Cr, Cu, Ni, Pb and Zn for each step of the BCR are given in Tables 2 and 3. Certified concentrations in the residue and the bulk sample, obtained after aqu regia extraction, are indicated in Pueyo et al. (2001) and Kubová et al. (2004).

2.3. Chemical analysis

Major and trace elements contents (Al, Ca, Fe, Mn, Cr, Cu, Ni, Pb and Zn) of leachates from the single extractions and the LP-extraction, as well as the residue, are carried out using inductively coupled plasma atomic emission spectrometry (ICP-AES; Varian Vista-MPX). The quality control of ICP-AES analyses is assessed by the analysis of blank reagents and calibration standards, prepared with commercially available solutions (Varian standard solutions). Accuracy of ICP-AES measurements is determined with various certified reference materials (Rousseau et al., 2009; Gabelle et al., 2012; Hamdoun et al., 2015a, 2015b). The detection limits (LD) range from 0.09 μg L⁻¹ to 8 μg L⁻¹, depending on the element.

Major element (Al, Ca, Fe and Mn) contents of supernatants from the BCR-extraction are analyzed by ICP-AES (Thermo IRIS INTREPID II XDL), while trace elements (Cr, Cu, Ni, Pb and Zn) are determined by ICP-MS (Agilent 7500) analyses. During the ICP-MS analyses, an internal standard, 115In/118Re of known concentration is added to all samples to correct the analyzer deviation. The detection limits vary around 0.2 mg L⁻¹ for ICP-AES and 0.01 μg L⁻¹ for ICP-MS, depending on the element. The accuracy of the ICP-MS measurements is checked using natural river water (SLRS-5) (Yeghicheyan et al., 2013) and the measurement errors are typically lower than 5%. All blanks were negligible.

Table 2
Chemical composition in mg·kg⁻¹ of dry soil of BCR 701 and recovery percentages calculated for both sequential extraction procedures.

<table>
<thead>
<tr>
<th>Element</th>
<th>Digestion of the bulk sample</th>
<th>Total concentration (mg·kg⁻¹)</th>
<th>% of recovery a</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td>See Table 1</td>
<td>This study</td>
<td>Kubová et al., 2004</td>
</tr>
<tr>
<td>Cr</td>
<td>378 ± 3</td>
<td>32</td>
<td>279 ± 9</td>
</tr>
<tr>
<td>Cu</td>
<td>309 ± 9</td>
<td>42</td>
<td>n.a.</td>
</tr>
<tr>
<td>Ni</td>
<td>123 ± 6</td>
<td>9</td>
<td>279 ± 9</td>
</tr>
<tr>
<td>Pb</td>
<td>164 ± 4</td>
<td>11</td>
<td>n.a.</td>
</tr>
<tr>
<td>Zn</td>
<td>518 ± 4</td>
<td>64</td>
<td>155 ± 4</td>
</tr>
<tr>
<td>Al</td>
<td>77,805 ± 3</td>
<td>64</td>
<td>476 ± 3</td>
</tr>
<tr>
<td>Ca</td>
<td>18,677 ± 3</td>
<td>504</td>
<td>71,470 ± 3</td>
</tr>
<tr>
<td>Fe</td>
<td>38,580 ± 3</td>
<td>518</td>
<td>n.a.</td>
</tr>
<tr>
<td>Mn</td>
<td>792 ± 3</td>
<td>689</td>
<td>n.a.</td>
</tr>
</tbody>
</table>

a % of recovery = sum of all fractions (mg·kg⁻¹)/concentration in the bulk (mg·kg⁻¹) × 100; n.a. = not available.
(less than $10^{-4}$ mg kg$^{-1}$).

### 3. Results and discussion

#### 3.1. Accuracy of the sequential extractions and of the bulk sample dissolution

Results on total metal concentrations on the bulk sample are given in Table 2. Most of the total concentrations are higher than the previous published data (Kubová et al., 2004), this may be due to a better digestion of clays or other silicate minerals, as suggested by the higher value of Al (see Table 2).

The accuracy of each step of the BCR-extraction procedure is additionally confirmed by comparison with Kubová et al. (2004) analyses and certified values (Pueyo et al., 2001) (see Table 3).

An internal check is secondly performed by comparing the bulk total concentrations of the target elements and the sum of the sequential fractions (see Table 2). The percentages of recovery of all studied elements, i.e. Al, Ca, Fe, Mn, Cr, Cu, Ni, Pb and Zn, range from 85% to 118% for both procedures, indicating the results were reliable considering the uncertainties (Table 2).

#### 3.2. Comparison of the labile fractions extracted by the four procedures

The so-called labile fraction is supposed to represent the environmentally available fraction of a specific element, if changes in the environment occur (Leleyter et al., 2012). In the leachates from the two sequential procedures and from the two single extractions, the so-called labile fraction is expressed by percentage values, which correspond to the ratio between the metal concentrations analyzed in the leachate and the total content determined in the bulk sample. Thus, for the sequential extractions, the percentage of the labile fraction, namely BCR-Flab and LP-Flab, are the sum of respectively the three (BCR-extraction) and the seven (LP-extraction) non-residual fractions divided by the total content determined in the bulk sample.

The Fig. 1 compares the percentages of labile fractions from the four extraction procedures for the target elements. Calculation of the standard deviation (SD) underlines that both sequential extraction procedures are very reproducible, since the values of SD range from 1 to 8%, depending on the elements (Fig. 1). The SD values of both single leaching procedures are correct but slightly higher since the values range from 1 to 12% depending on the element, involving a heterogeneity of the distribution of this element in the sampled aliquots (Fig. 1).

Among the major elements, Al is poorly mobilized by the four procedures ($\leq 12\%$), indicating that the major part of Al remains trapped in the residual fraction since the reagents do not affect the alumino-silicate

---

**Table 3**

Comparison between the certified concentrations of the BCR-701 reference material for Cd, Cr, Cu, Ni, Pb and Zn (mg kg$^{-1}$) (Pueyo et al., 2001) and the concentrations measured in this study and in Kubová et al. (2004), from the BCR-extraction. (na: not available).

<table>
<thead>
<tr>
<th>Element</th>
<th>Acid-soluble (mg kg$^{-1}$)</th>
<th>Reducible (mg kg$^{-1}$)</th>
<th>Oxidizable (mg kg$^{-1}$)</th>
</tr>
</thead>
<tbody>
<tr>
<td>Cr</td>
<td>2.6 ± 0.5 na</td>
<td>2.3 ± 0.2 na</td>
<td>36 ± 7 na</td>
</tr>
<tr>
<td>Cu</td>
<td>46 ± 6 51 ± 1</td>
<td>49 ± 2 19 ± 2</td>
<td>129 ± 9 118 ± 2</td>
</tr>
<tr>
<td>Ni</td>
<td>15 ± 2 na</td>
<td>15 ± 1 na</td>
<td>26 ± 5 na</td>
</tr>
<tr>
<td>Pb</td>
<td>3 ± 0.5 3 ± 0.5</td>
<td>3 ± 0.5 3 ± 0.5</td>
<td>120 ± 8 128 ± 1</td>
</tr>
<tr>
<td>Zn</td>
<td>197 ± 18 200 ± 1</td>
<td>205 ± 6 194 ± 3</td>
<td>118 ± 2 114 ± 5</td>
</tr>
<tr>
<td>Al</td>
<td>158 ± 1 182 ± 2 na</td>
<td>na 2242 ± 1 162</td>
<td>3526 ± 46 na</td>
</tr>
<tr>
<td>Fe</td>
<td>54 ± 1 71 ± 1 na</td>
<td>na 4432 ± 35 7698 ± 106</td>
<td>7698 ± 106 na</td>
</tr>
<tr>
<td>Ca</td>
<td>7752 ± 119 na</td>
<td>na 1990 ± 56 7698 ± 106</td>
<td>na 1990 ± 56 7698 ± 106</td>
</tr>
<tr>
<td>Mn</td>
<td>174 ± 2 170 ± 1 na</td>
<td>na 117 ± 3 125 ± 2 na</td>
<td>26 ± 1 23 ± 1 na</td>
</tr>
</tbody>
</table>

**Fig. 1.** Percentage of labile metal from single (HCl-Flab or EDTA-Flab) and sequential (BCR-Flab or LP-Flab) extractions. Uncertainties are calculated from the triplicates.
3.3. Comparison between the geochemical partitioning of elements determined from both sequential extraction procedures

In order to investigate the differences found between the total labile fractions of both sequential extraction procedures, a comparison of each extracted non-residual fraction is performed. However, when comparing both sequential extraction procedures, equivalent fractions need to be defined as the BCR comprises three steps compared to seven steps in the LP-extraction (see Table 1):

- The BCR-F1 fraction is the first step in the BCR procedure and solubilize the acido-soluble fraction. It can be compared to the first three fractions of the LP-extraction, which solubilize the water-soluble fraction (F1a), the exchangeable fraction (F1b) and then the soluble acido fraction (F1c).
- In the same way, the BCR-F2 fraction, which solubilize the reducible fraction, can be compared to the sum of the three following LP-fractions, which solubilize the Mn-oxides (F2a), the amorphous iron oxides fraction (F2b) and the crystalline iron oxides fraction (F2c).
- The BCR-F3 and LP-F3 fractions can be compared as these two extractions solubilize the oxideable fraction of the sediment or soil.

3.3.1. Acido-soluble fraction

The both acido-soluble fractions (BCR-F1) and (LP-F1(a + b + c)) are reported in Figs. 2 and 3 and Tables 4 and 5. The standard deviations range from 0 to 4%, depending on the considered element (Tables 4 and 5). It underlines the reproducibility of the acido-soluble step(s) whatever the considered procedure. Three groups of elements are noticed:

- Al, Cr and Fe which are element with no bound with the acido-soluble fraction characterized by the very low F1 extractability (<3%).
- elements with similar extracted percentages by the both procedures Mn (22 and 21%), Ni (11 and 8%) and Zn (34 and 31%) respectively for BCR- and LP-F1 fractions.
- elements which are better extracted by the LP-extraction than by the BCR-extraction Ca (47 and 42%), Cu (25 and 15%) and Pb (22 and 2%) respectively for LP and BCR-F1 extractions.

Generally, the exchangeable step of a sequential extraction procedure is based on the replacement of an exchangeable cation (which is adsorbed onto solid material due to permanent structural charges, so generally located in the interlayer spaces of clay minerals), by another one from a chemical reactive, which has to be a polyvalent cation to optimize this exchange. In the BCR, the cation used in BCR-F1 is the oxonium ion (H3O+), which is a really poor exchangeable cation contrary to the bivalent ions Mg2+, used in the LP-procedure. Thus, it is not surprising that the BCR exchangeable fraction is underestimated. Moreover, in this procedure, the water soluble, the exchangeable and the acido-soluble fractions are extracted in one step with a lower concentration of reagent compared to LP-extraction. The dissolution of the carbonate fraction is probably not complete in BCR, as suggested by the relative decrease of the percentage of Ca extracted in BCR (42%) compared to LP-extraction (47%). Sulkowski and Hirner, (2006) and Dodd et al. (2000) also concluded that the first step of BCR failed to completely leach the carbonate fraction of high carbonate content samples. Then the lower extraction of Ca by LP extraction may be due to the removal of metals present in forms other than exchangeable or carbonate-bound. Indeed, Hanahan (2004) found that 1 M Na-acetate, pH 5, released metal associated with hydroxide minerals.
Fig. 2. Percentage of labile metal from sequential BCR (F1 to F3) and LP (F1 to F3) extractions.

Fig. 3. Percentage of metal bound to acido-soluble fraction from BCR-extraction (BCR-F1) and LP-extraction (LP-F1a: water soluble fraction, LP-F1b: exchangeable fraction, LP-F1c: real acido-soluble fraction).

Table 4
Percentage of labile metal from acido-soluble, reducible and oxydizable fractions from the sequential BCR extraction (F1 to F3). Uncertainties are calculated from the triplicates.

<table>
<thead>
<tr>
<th>%a</th>
<th>Al</th>
<th>Ca</th>
<th>Fe</th>
<th>Mn</th>
<th>Cr</th>
<th>Cu</th>
<th>Ni</th>
<th>Pb</th>
<th>Zn</th>
</tr>
</thead>
<tbody>
<tr>
<td>BCR-F1</td>
<td>0.2</td>
<td>42</td>
<td>0.1</td>
<td>22</td>
<td>0.5</td>
<td>15</td>
<td>11</td>
<td>2</td>
<td>34</td>
</tr>
<tr>
<td></td>
<td>0.001</td>
<td>0.6</td>
<td>0.004</td>
<td>0.2</td>
<td>0.01</td>
<td>1.9</td>
<td>0.3</td>
<td>0.2</td>
<td>0.8</td>
</tr>
<tr>
<td>BCR-F2</td>
<td>3</td>
<td>11</td>
<td>12</td>
<td>15</td>
<td>8</td>
<td>42</td>
<td>18</td>
<td>73</td>
<td>19</td>
</tr>
<tr>
<td></td>
<td>0.2</td>
<td>0.2</td>
<td>0.1</td>
<td>0.4</td>
<td>0.1</td>
<td>3</td>
<td>0.3</td>
<td>5</td>
<td>1</td>
</tr>
<tr>
<td>BCR-F3</td>
<td>1</td>
<td>1</td>
<td>2</td>
<td>3</td>
<td>38</td>
<td>17</td>
<td>12</td>
<td>5</td>
<td>9</td>
</tr>
<tr>
<td></td>
<td>0.1</td>
<td>0.03</td>
<td>0.1</td>
<td>0.1</td>
<td>0.9</td>
<td>1</td>
<td>0.6</td>
<td>0.5</td>
<td>0.5</td>
</tr>
</tbody>
</table>

a Average in bold and standard deviations in italics.

Table 5
Percentage of labile metal from acido-soluble, reducible and oxydizable fractions from the sequential LP extraction (F1 to F3). Uncertainties are calculated from the triplicates.

<table>
<thead>
<tr>
<th>%a</th>
<th>Al</th>
<th>Ca</th>
<th>Fe</th>
<th>Mn</th>
<th>Cr</th>
<th>Cu</th>
<th>Ni</th>
<th>Pb</th>
<th>Zn</th>
</tr>
</thead>
<tbody>
<tr>
<td>LP-F1</td>
<td>0.2</td>
<td>47</td>
<td>1</td>
<td>21</td>
<td>3</td>
<td>25</td>
<td>8</td>
<td>22</td>
<td>31</td>
</tr>
<tr>
<td></td>
<td>0.01</td>
<td>4</td>
<td>0.0</td>
<td>3</td>
<td>0.1</td>
<td>1</td>
<td>0.5</td>
<td>1</td>
<td>1</td>
</tr>
<tr>
<td>LP-F2</td>
<td>9</td>
<td>2</td>
<td>24</td>
<td>36</td>
<td>46</td>
<td>67</td>
<td>47</td>
<td>6</td>
<td>50</td>
</tr>
<tr>
<td></td>
<td>1</td>
<td>0.2</td>
<td>2</td>
<td>3</td>
<td>5</td>
<td>6</td>
<td>4</td>
<td>2</td>
<td>4</td>
</tr>
<tr>
<td>LP-F3</td>
<td>3</td>
<td>&lt;LD</td>
<td>7</td>
<td>4</td>
<td>16</td>
<td>8</td>
<td>8</td>
<td>1</td>
<td>10</td>
</tr>
<tr>
<td></td>
<td>0.3</td>
<td>0.2</td>
<td>0.4</td>
<td>1</td>
<td>0.8</td>
<td>0.5</td>
<td>0.5</td>
<td>0.5</td>
<td>0.5</td>
</tr>
</tbody>
</table>

a Average in bold and standard deviations in italics.
the considered element (Tables 4 and 5). It underlines the reproducibility of the reducible step(s) whatever the considered procedures.

The percentages extracted by the LP-extraction are, on average, 2 to 6 times higher than those from the BCR-procedure, for all elements, except for Ca and Pb. The LP-F2 percentage of Fe and Mn extracted are respectively 24 and 36% and are significantly higher compared to 12 and 15% extracted by F2-BCR. This result (agreed with Anju and Banerjee, 2010 and Caraballo et al., 2018) suggests a lower efficiency of the BCR-procedure in the dissolution of Mn- and Fe-oxides.

The sum of Fe and Mn extracted at the Mn-oxides (F2a) and the amorphous Fe-oxides (F2b) fraction in the LP-extraction are 16 and 28%, respectively. The percentage of amorphous Fe extracted is comparable to that obtained from the BCR, but a lower dissolution of the Mn-oxides is highlighted. We can conclude that BCR-procedure is not able to solubilize neither the totality of the manganese oxides (the quantity of solubilized manganese is even less important than that mineralized after EDTA-extraction) nor the crystalline iron oxides. The infective dissolution of iron oxides was already observed when applying BCR-extraction procedure (Leermakers et al., 2019). These results can be explained by the fact that the reducible step from BCR is realized at the room temperature. Crystalline Fe-oxides are only poorly (or not) dissolved, even considering the low pH of this step. Moreover, hydroxylation is known to be a powerful reducing agent, a perfect reagent to reduce insoluble MnIV to soluble MnII, but this reagent is not efficient enough to mineralize iron oxides, like goethite (1) or hematite (2), characterized by a lower standard potential (respectively E° = 0.92 and 0.86 V/ENH).

Leleeyter and Probst (1999) measured the percentage of leached Fe during an oxide leaching procedure (NH2OH⋅HCl 0.1 M, pH = 3.5, 30 min) performed on four different iron oxides: a synthetic hematite black powder from Schering A.G. Berlin, a natural goethite ochre powder from Atlantis II, carrot 1032 VIII 1-6, a natural mix hematite and goethite orange powder from Atlantis II, carrot 1032 XI 75-80 and a synthetic hematite red powder from Merck. For all these oxides, the percentage is equal to zero, confirming the low efficiency of this reagent to dissolve Fe-oxides. However, the reducing power of this reactant can be enhanced in more extreme conditions, such as heating or low pH. However such conditions provoke the partial dissolution of clay minerals (Robbins et al., 1982) then a potential overestimation of the extracted percentages for the reducible fraction.

Thus, the high percentages of Cr, Cu, Ni and Zn extracted at the reducible step of LP-extraction can be probably explained by the combination of the reagents used and by the division of the reducible fraction into three distinct steps, allowing the increase of the dissolution efficiency of amorphous and crystalline oxides. Indeed, Leleyter and Probst (1999) found that the leached Fe percentage by the reducible step of LP-extraction extracted from 73 to 97% of the synthetic iron oxides. Additionally, Davidson et al. (2004) found that the substitution of the hydroxylation chloride by ammonium oxalate at the step 2 of the BCR step allowed an increase of the extracted Fe- and Cu- and also Zn-contents. It was explained by the ability of ammonium oxalate to liberate iron and related elements from mineralogical phases which would be classed, under the BCR scheme, as part of the residual fraction. On the contrary, the BCR probably leads to an underestimation of elements associated with the reducible phase and a potential overestimation of the residue.

Contrary to the others elements, Ca and Pb are better extracted in the reducible fraction of BCR than LP-extraction. As all the labile fractions, BCR-Flab, LP-Flab HCl-Flab and EDTA-Flab give quite similar results for calcium extraction (53, 50, 50 and 49% respectively), it can be assumed that a better calcium extraction during the 3 steps of the LP-F1 (than BCR-F1), explains a lower extraction of calcium during LP-F2. On the contrary, the BCR extracted over 40% of lead compared with the LP-extraction, but also compared with both single leaching procedures (HCl and EDTA). The particular case of lead will be discussed further in this study.

3.3.3. Oxidizable fraction

The percentages extracted by the both procedures (BCR-F3 compared to LP-F3) are given in Tables 4 and 5 and Fig. 2. The standards deviation ranges from 0,03 to 1%, which underlines once again the reproducibility of the oxidizable step. The percentages of Cr, Cu, Ni and Pb, leached from the oxidizable phase, are significantly higher in the BCR than in the LP-extraction.

Sahuquillo et al. (2003) highlighted a decrease of the oxidizable fraction of the reference material CRM-601 between the first (from Quevauviller et al., 1994) and the modified (from Rauret et al., 1999) standardized sequential procedure, for Cr, Cu, Pb and into a lesser extent, Ni. Indeed, in the modified procedure, the decrease of the oxidizable fraction is associated with an increase of the reducible one, involving two assumptions. The first one is the decrease of the red-adsorption phenomena in the reducible step, previously occurring in the procedure from Quevauviller et al. (1994). These red-adsorption phenomena would lead to an overestimation of the oxidizable fraction of this procedure. Conversely, the second assumption is the increase of the specificity of the extraction in the BCR, thus limiting the subsequent dissolution of the Fe-oxides in the oxidizable phase of the procedure.
from Quevauviller et al. (1994). Following this assumption, this fraction was previously only partially digested during the reducible step.

A similar trend was observed in the present study, i.e. an increase of the reducible fraction in the LP-extraction associated with a decrease of the oxidizable one, compared to the BCR, for the same elements. As an example, the extracted percentage of Cr and Cu of the reducible fraction are 46% and 67% then 8% and 42%, in the LP-extraction then BCR respectively. On the contrary, the values are 16% and 8% then 38% and 17% in the oxidizable fraction for the LP then BCR-extractions respectively. Assuming that the low pH (1.5) of the reducible fraction of the BCR prevent re-adsorption processes involving further overestimation of the following step, one explanation could be that the Fe-oxides, only partially dissolved in the reducible step of the BCR, are further dissolved in the oxidizable one. The comparison between the sum of the oxidizable and the reducible fractions of both procedures are similar and thus let us confident with this assumption (Fig. 2).

Alternatively, the increase of the extracted percentage of Cu during the reducible fraction of the LP-extraction could be due to an overestimation of the reducible fraction. Indeed, copper oxalate complex formation is relatively easy. Thus, these complexes could remove copper from the oxidizable phases (Slavek and Pickering, 1986; Benitez and Dubois, 1999; Davidson et al., 2004). Other assumptions could involve (1) a re-adsorption phenomena occurring in the oxidizable step of the LP-extraction, (2) a lower ability of this procedure to extract the elements located in this fraction, according to Fernández Alborés et al. (2000), or (3) a premature extraction of organically-bound metals from the exchangeable fraction, as found by Gómez-Arizá et al. (2000) for the four step Tessier procedure (Tessier et al., 1979). This third hypothesis could be supported by the higher percentage of Cu in the LP-F1 (25%) compared to the BCR-F1 (15%).

3.4. Contribution of single and sequential leaching procedures on the understanding of labile Pb behavior

Contrary to the other elements, lead shows a preferential extraction by the BCR (Fig. 1). Thus, lead re-adsorption and/or co-precipitation phenomena might occurred in the other procedures. Davidson et al. (2004) used 0.2 mol L\(^{-1}\) of ammonium oxalate instead of 0.5 mol L\(^{-1}\) hydroxylammonium hydrochloride, for an identical pH value of 1.5. It decreased the proportions of Ca and Pb released, for the reducible step performed on the certified lake sediment CRM-601. According to these authors, a rapid precipitation of analytes oxalates from the solution is the most likely assumption explaining the decrease found in the LP-extraction. Additionally, Sahuquillo et al. (1999) also found that the replacement of hydroxylammonium chloride by ammonium hydrogen oxalate and oxalic acid induced the precipitation of insoluble Pb salts, in particular in the presence of Ca (due to co-precipitation phenomenon). Thus, similar explanation could be expected in our study. The relatively low solubility of Pb oxalate (K\(_s\) = 1.7 \times 10\(^{-10}\)) and the high amount of lead present in the extract, as deduced from the BCR procedure, could induce the precipitation of lead oxalate from solution (Pickering, 1986; Benitez and Dubois, 1999; Gleyzes et al., 2002; Davidson et al., 2004). It is consistent with the fact that no Ca is found in the extracts corresponding to the oxalate steps, i.e. LP-F2b and LP-F2c (see Fig. 4).

The total labile fraction of Pb is also higher in the BCR than in both single leach procedures. In the case of 0.5 mol L\(^{-1}\) HCl, the low percentage of Pb extracted could be explained by the Pb re-adsorption and re-distribution processes. Indeed, PbCl\(_2\) precipitation phenomenon (K\(_s\) = 1.8 \times 10\(^{-5}\)) were already observed by Wei et al. (2005), by XAS analysis. They reported that, during a 0.1 mol L\(^{-1}\) HCl extraction, the main part of lead is converted into precipitated PbCl\(_2\). They concluded that estimating the level of metals uptake by plant material using 0.1 mol L\(^{-1}\) HCl, can underestimate Pb uptake because lead cations readily reacted with chloride to form PbCl\(_2\) precipitates that are only poorly soluble in dilute acid or water. In the case of 0.05 mol L\(^{-1}\) EDTA leach procedure, Kim et al. (2003) found that major cations like Fe and trace elements like Cu and Zn potentially compete with lead for EDTA ligand sites. Moreover, according to this study, the occlusion of Pb on Fe oxides and the type of Pb species can affect the Pb extraction efficiency. On the contrary, Leleyter et al. (2012) assumed that 0.05 mol L\(^{-1}\) EDTA extraction was the best suited method to estimate the Pb mobility, compared to HCl and LP-extraction for various soils and sediments.

Alternatively, the better extraction of Pb by the reducible step of BCR could be explained by the specific physico-chemical parameters of its reducible step. The increase of the concentration of the hydroxylammonium hydrochloride, from 0.1 mol L\(^{-1}\) (in the original standardized BCR procedure) to 0.5 mol L\(^{-1}\) (in the modified one), lead to an increase up to 30% of the percentage of Pb extracted (Rauret et al., 1999; Sahuquillo et al., 2003). On the contrary, the concentration of the reactants used in the different steps of the reducible fraction of the LP-extraction is 0.2 mol L\(^{-1}\) only. Moreover, the pH value of the reducible step is 1.5 for the BCR and 3.5 to 2.3 for the LP-extraction; this lower pH could also explain better extractability of Pb by BCR, as suggested by Sahuquillo et al. (1999), since the extractability of lead is sensitive to a decrease of pH.

4. Conclusions

The aim of this study was to compare the potential availability and the partitioning of Cr, Cu, Zn, Ni and Pb between four leaching procedures commonly found in the literature, i.e. two single (0.5 mol L\(^{-1}\) HCl and 0.05 mol L\(^{-1}\) EDTA) and two sequential (from Rauret et al., 1999 and from Leleyter and Probst, 1999) extractions. The procedures were all performed on the reference lake sediment BCR-701, commonly used to assess the accuracy of the modified standardized procedure of Rauret et al. (1999). Major elements like Al, Ca, Fe and Mn were also studied in order to investigate the selectivity of the four used procedures.

Similar percentages of Cr, Cu, Pb and Zn are extracted by both single leach procedures, but the dissolved phases are probably not the same, as suggested by the difference between the percentages of Al and Mn extracted. Thus, 0.5 mol L\(^{-1}\) HCl could be more efficient than 0.05 mol L\(^{-1}\) EDTA to remove trace elements from the reducible phase. The comparison between the single leaching procedures and the sum of various steps of the two sequential extraction procedures indicate that all the operationally defined fractions are impacted by dilute HCl and 0.05 mol L\(^{-1}\) EDTA extractions. Thus, it is difficult to associate the amount of trace elements extracted by one or the other single reactant, to one particular mineralogical fraction. As a whole, the LP-extraction appears more extractive and efficient than the standardized BCR procedure for all elements (excepted lead) probably due to a better extraction of the exchangeable fraction, to a better dissolution of the acido-soluble phase and a better dissolution of the amorphous and crystalline Fe-oxides.

On the contrary, BCR-extraction is more efficient to extract Pb (i.e. 80%) compared to the others three procedures, that show similar amount of extracted Pb (ranging from 24 to 29%). Two assumptions could be proposed. The first one involves co-precipitation and/or low-extraction processes occurring in the three others procedures. The second one involves the particular physico-chemical properties of the reducible phase of BCR (i.e. low pH and strong reactant concentration). These conditions could induce an extraction of Pb previously retained in the residual phase, and thus an overestimation of the real Pb availability.

Finally, Ca and, in a lesser extent Cr, are the only elements showing close availabilities by the four procedures. LP-extraction seems the more suitable method to estimate the potential availability of Cu, Ni and Zn whereas BCR could be more adapted to estimate the Pb availability. However, the choice of the procedure and of the reactants must be previously deduced from the physico-chemical properties of the sample and the aim of the study. A mineralogical characterization of the residual phase could help to understand if co-precipitation phenomena have occurred in the procedure of Leleyter and Probst (1999) or during...
the 0.5 mol·L⁻¹ HCl single leach, explaining the difference of the Pb lability fractions between the procedures.

Declaration of competing interest
The authors certify that they have no affiliations with or involvement in any organization or entity with any financial interest (such as hono-
raria; educational grants; participation in speakers’ bureaus; member-
ship, employment, consultancies, stock ownership, or other equity interest; and expert testimony or patent-licensing arrangements), or non-financial interest (such as personal or professional relationships, affiliations, knowledge or beliefs) in the subject matter or materials discussed in this manuscript.

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References
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