

Chemical variations of loess from the Chinese Loess Plateau and its implications

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Keywords:	Upper continental crust, loess, Chinese Loess Plateau, elemental behaviors, carbonate
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Highlights

- Fluid-soluble elements are conserved during sedimentary processes to produce loess from Chinese Loess Plateau.
- Highlight the significant contribution of carbonate for estimates of average composition of upper continental crust.
- Important updates to the loess composition and average composition of the upper continental crust.

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26 Abstract

Chinese Loess Plateau (CLP) is the largest loess deposit on the Earth with expansive surface-exposed source rocks of varying origin, age, and history through thorough mixing during long-distance transport. We present elemental abundances on representative loess and paleosol samples from seven classic sections of the CLP. Most elements, including water-soluble elements (e.g., Rb and Cs), show significant correlations with La or Al₂O₃. These correlations indicate the conservation of these elements during weathering, transport, and deposition hosted or absorbed in particle minerals (e.g., mica, K-feldspar, and clay minerals). These new observations allow the use of La/X ("X" being the element of interest), and using the widely accepted La content of 31 ppm in the model upper continental crust (UCC) to refine the UCC composition. The results show higher Cs (Cs = 6.7 ± 1.2 ppm), lower transition metals, Ba, and Ga. Given the high CaO and presence of carbonate in UCC rocks of both vast western China (the primary source for the CLP) and eastern China, we propose that these updates on the element abundances of the CLP loess represent a possibly improved model for the carbonate-bearing UCC.

42 Keywords: Upper continental crust; loess; Chinese Loess Plateau; elemental
43 behaviors; carbonate

1. Introduction

The scientific significance of upper continental crust (UCC) on which we live is self-evident as it represents the surficial end-product of chemical differentiation of the Earth over its long and complex history. As the starting point to understand the formation of continental crust, geochemists have been endeavoring to estimate the average composition of continental crust, especially the UCC, using various methods over the past century. Two approaches have been commonly taken: (1) weighted averages of bedrock compositions through systematic large-scale sampling (Shaw et al., 1967; Gao et al., 1998) and (2) analyzing average compositions of fine-grained sediments or sedimentary rocks (e.g., shales) in the UCC (Condie, 1993; Chauvel et al., 2014). These methods provide good constraints on the abundances of major elements and water-insoluble minor and trace elements of the UCC. However, the abundances of transition metals and water-soluble elements have not been well constrained because of preferential/differential chemical weathering and limited analyses.

Loess, formed as the result of wind-blown silt-size particle deposition, occupies ~ 10% of the Earth's land surface (Fig. 1; Pye, 1995; Taylor et al., 1983). Because it was formed in arid and/or cold climatic conditions (i.e., peridesert loess and periglacial loess; Pye, 1995), loess has experienced physical weathering with limited chemical weathering during its transport and deposition (Gallet et al., 1998; Peucker-Ehernbrink and Jahn, 2001). Hence, loess best reflects source rock compositions with no or rather limited element fractionations. Furthermore, global loess shows similar patterns and ratios of rare earth elements (REEs) and Th (e.g., Taylor et al., 1983),
which are also comparable to those of the UCC estimated by using other approaches
(Fig. 2). Hence, loess can be taken as a natural mixture of materials eroded from the
expansive regions of the surficial crust (Chauvel et al., 2014; Gallet et al., 1998;
Taylor et al., 1983).

Previous studies, however, show that the contents of water-soluble trace elements (e.g., K, and Ba) in loess are either uncorrelated or poorly correlated with water-insoluble elements, such as La (e.g., Rudnick and Gao, 2003 and references therein), suggesting that water-soluble elements may have been differentially leached during weathering. Consequently, the contents of water-insoluble trace elements (e.g., Nb, Ta, and Th) in loess (Fig. 2a,b) are thus used to estimate the contents of water-insoluble elements in the UCC. Some other studies, however, suggest that some of the "weathering signatures" may in fact be inherited from eroded bedrocks, recording previous weathering events (e.g., Peucker-Ehernbrink and Jahn, 2001). Moreover, recent studies have also shown that loess composition can vary with varying provenances in space and time (Chen et al., 2007; Chen and Li, 2013; Nie et al., 2014; Sun et al., 2010; Sun and Zhu, 2010; Xiao et al., 2012). In addition, eroded materials may have experienced mineral sorting, which can lead to the fractionation of heavy minerals and their hosted elements (Taylor and McLennan, 1995).

Whether a chemical element in loess can be used to infer its content in the
average UCC thus depends on the source inheritance and mixing processes.
Therefore, a clear understanding of elemental behaviors in all aspects of sedimentary

processes to produce loess is required for better constraining the average composition of the UCC using the loess composition. Compared to periglacial loess (e.g., western Europe loess, Argentinian), dust in peridesert loess was derived from different source rocks of a larger area with longer distance transport and more effective mixing (e.g., Chauvel et al., 2014). Among these peridesert loess deposits, the Chinese Loess Plateau (CLP) is the thickest and most extensive loess deposit on the Earth (e.g., Liu and Ding, 1998) with an exposure area in excess of c. 635,280 km² and thickness of up to c. 505 m locally (Li et al., 2020). Here, we present chemical compositions of loess samples from seven representative sections of the CLP (Fig. 1b-e), and discuss the controls on the behaviors of various elements during sedimentary processes from weathering to erosion and to deposition. This study showed that the CLP loess is an excellent candidate for estimating the average composition of the carbonate-bearing UCC for most elements, including water-soluble elements (e.g., K, Rb, Cs, Ba).

2. Samples and methods

102 2.1 The geological setting of the CLP and sampling sections

Aeolian dusts of the CLP are interpreted to have derived from vast arid lands upwind to the west and northwest (e.g., the Taklamakan Desert in western China, the Qaidam Gobi-Desert on the northern Tibetan Plateau, and the Badain Jaran and Tengger deserts in northern China) and originated from erosion of the surrounding mountain ranges (i.e., the Altyn mountains, Qilian mountains, and Kunlun mountains on the north edge of the Northern Tibetan Plateau, and the Tianshan mountains and

Gobi Altay mountains from the Central Asian Orogenic belt as shown in Fig. 1b; Chen et al., 2007; Chen and Li, 2013; Xiao et al., 2012). Because of the Tibetan Plateau uplift and northern hemisphere glaciation (e.g., An et al., 2001; Sun et al., 2010, 2020; Sun and Zhu, 2010), vast areas of bedrocks have been exposed for erosion to supply aeolian dusts to the CLP. Varied provenances and long-distance transport in combination facilitate effective dust mixing compared to periglacial loess (e.g., Western Europe loess; Chauvel et al., 2014; Sauzéat et al., 2015).

We sampled seven classic sections of the CLP in this study: Luochuan (6 loess samples and 5 paleosol samples), Yan'an (2 loess samples), Hukou (5 loess samples), Lingtai (4 loess samples), Lantian (3 loess samples and 2 paleosol samples), Jiuzhoutai (10 loess samples), and Qin'an (5 loess samples; Fig.1b-e). All these CLP sections are Quaternary deposits of < c. 2.6 Ma, but the Qin'an section that is significantly older, deposited at $\sim 22-6.2$ Ma as evidenced by paleomagnetic data and fossils (e.g., Guo et al., 2002). Bulk-rock loess samples were analyzed in this study to avoid artificial fractionation caused by varying grain sizes, which reflect varied sources and complex sedimentary histories.

2.2 Analytical methods

The loess samples were carefully hand-crushed in an agate mortar into powders in a clean environment for bulk-rock analysis to avoid potential contaminations. The samples were analyzed in the Laboratory of Ocean Lithosphere and Mantle Dynamics (LOLMD) in the Institute of Oceanology, the Chinese Academy of Sciences.

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Inductively coupled plasma optical emission spectrometer (ICP-OES, Agilent 5100) and inductively coupled plasma mass spectrometer (ICP-MS, Agilent 7900) were used for analyzing major and trace elements, respectively. Strontium isotope compositions were measured on a multi-collector inductively coupled plasma mass spectrometer (MC-ICP-MS, Nu II). The analytical data of chemical compositions and Sr isotope compositions are given in Supplementary Table 1-3, and detailed information for analytical precision and accuracy are given in Supplementary File A. For major elements, each dried sample powder of 50 mg mixed with 250 mg lithium borate flux (LiBO₄) was melted in a platinum crucible at 1050°C for 0.5 hour in a muffle furnace, followed by heating over a Bunsen burner at 1000°C. Then, the melted droplet was immediately dropped into c. 50 mL 5% HNO₃ solution and diluted to 100 mL with Mili-Q water (Kong et al., 2019). Our repeated analyses of United States Geological Survey (USGS) rock standards (BHVO-2, AGV-2, and GSP-2) agree with recommended values, better than 5% for accuracy and better than 3% for precision (Supplementary File A). For loss on ignition (LOI), ~ 500 mg powder for each sample was heated in a muffle furnace at 1000°C for 0.5 hours with the calculated weight loss as the LOI.

For trace element analysis, 50 mg powder for each sample was dissolved in antiaqua regia (HNO₃ : HCl = 3:1) and HF in a high-pressure bomb (a Teflon beaker in a
stainless steel jacket) for 15 hours, followed by 2 hours re-digestion with 20% HNO₃,
following Chen et al. (2017). Then, each sample solution was diluted to 100g (with
dilution factor of 2000) in 2% HNO₃ for analysis. During analysis, ICP-MS was

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152	settled in no gas mode instead of using collision mode, and calibration was performed
153	based on five solutions (1, 10, 25, 50 and 100 ng/mL for all the analyzed elements)
154	acquired from multi-element calibration standard solutions (Agilent Technologies,
155	Tokyo, Japan) with a blank. One replicate sample was analyzed for every ten samples,
156	and a given lab-mixed solution was analyzed every four samples to monitor
157	instrumental drift. USGS standards (BCR-2, AGV-2, and GSP-2) were analyzed as
158	unknowns to determine accuracy and precision (Supplementary File A). Accuracy,
159	indicated by RE between analyzed values of USGS standards and their recommended
160	values, is generally better than 10% for most trace elements (except for BCR-2 with
161	12% RE for Ni and Cu, and 15% for Sn), with many elements agreeing with the
162	reference values within 5% (Supplementary File A). Precision, indicated by replicate
163	analyses, is within 5% for most trace elements (Supplementary File A).

For Sr isotope analysis, 50 mg sample powder was decomposed in a high-164 pressure bomb by using HNO₃ + HCl + HF at 190°C for 15 hours, followed by re-165 digestion with 2 mL 3N HNO₃ for 2 hours. Then, sample solutions were loaded onto 166 Sr-spec resin columns for Sr separation, and dry plasma mode with Aridus II was 167 used for analysis. As the relative abundances of ⁸³Kr and ⁸⁶Kr are constant and no 168 interference for ⁸³Kr, ⁸⁶Kr was calculated based on measured ⁸³Kr. The intensity of 169 ⁸⁶Sr was acquired by reducing ⁸⁶Kr from the total measured isotope at mass 86. The 170 measured 87 Sr/ 86 Sr ratios were normalized to 86 Sr/ 88 Sr = 0.11940 to remove the effect 171 of instrumental mass fractionation. NBS-987 was analyzed bracketing every four 172 samples to monitor the instrument drift during the analysis. The repeated 173

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measurements of 87 Sr/ 86 Sr ratios of NBS-987 is 0.710244 ± 0.000009 (n = 9, 2 σ), consistent with the reference value (0.710243 - 0.710250; http://georem.mpch-mainz.gwdg.de/sample query pref.asp). Analysis of BCR-2 gives ⁸⁷Sr/⁸⁶Sr = 0.705336 ± 0.000007 , which agrees well with the recommended value of $0.70492 \pm$ 0.00055.

For petrographic studies, mineral phases in thin sections were identified by using focused ion beam (FIB) - scanning electronic microscopy (SEM, Zeiss Gemini 2 crossbeam 550) with energy dispersive spectrometer (EDS, Bruker QUANTAX) at China University of Petroleum (East China). To determine mineral modal ıcaı. abundances, advanced mineral identification and characterization system (AMICS) software was used.

3. Results

3.1 Mineral counting results

The back-scattered images and mineral modal abundances acquired using SEM-EDS with AMICS are shown in Fig. 3. The mineralogy of the CLP loess and paleosol is primarily composed of quartz + feldspar (albite, K-feldspar and albite-anorthite) + carbonate + micas (both biotite and muscovite) + epidote with allanite + clay minerals (illite, kaolinite and chlorite). Heavy minerals include amphibole, augite, garnet, rutile, titanite, apatite and Fe-oxides. These mineral assemblages we obtained in the CLP loess confirm previous petrographic studies, in which all these minerals are reported either as single particles or parts of rock fragments in the CLP loess (Jeong et

al., 2008, 2011). Ca-carbonate minerals are common in the loess, and can reach up to *c*. 40 wt.% (16QA-02) in the Qin'an section. Carbonate mainly occurs as detrital
grains, evidenced by its morphology as single angular particles or as parts of rock
fragments, together with some secondary carbonate (also see Supplementary File B).
In comparison, the paleosols have little carbonate and much less epidotite epidote
group minerals, but more clays (Fig. 3).

3.2 Major elements

For both the CLP loess and paleosol samples, Al₂O₃ shows negative correlations with CaO and positive correlations with TiO₂-FeOT-K₂O-MnO (Fig. 4a-e). CaO and LOI are positive correlated (Fig. 4f), reflecting that the determined LOI largely results from carbonate decomposition. Compared with the Quaternary loess samples, the Qin'an loess samples have higher CaO-LOI (up to 22.40 wt.% CaO and 21.27 wt.% LOI) and lower contents of other major elements, while paleosol samples have obviously lower CaO-LOI-Na₂O-P₂O₅ and higher SiO₂-TiO₂-Al₂O₃-Fe₂O₃-K₂O (Fig. 4 and Supplementary Table 1).

3.3 Trace elements

All the CLP loess and paleosol samples show a uniform trace element pattern (Fig. 2c,d), i.e., elevated Ba–Rb–Cs–Th–U–Pb–LREEs, low Nb–Ta–Sr, a clear negative Eu anomaly (i.e., $Eu_N/Eu* = Eu_N/(Sm*Gd)^{1/2} \approx 0.6$), and a flat HREE pattern. This pattern is similar to that of loess samples elsewhere reported in the

literature, except that we do not see large positive Zr-Hf anomalies in the CLP loess and paleosol samples (Fig. 2c,d). Most analyzed elements of the CLP Quaternary loess samples show positive correlations with La (Figs. 5a-k&6a-f), including both water-soluble elements (e.g., Rb-Cs-Ba-Pb) and insoluble elements (i.e., REE and Ti-Nb-Ta-Th). However, it lacks significant correlations with Zr or Hf with the increase of La (Fig. 51). Strontium shows a positive correlation with CaO but not with others. In addition, Molybdenum and U show scattered positive correlations (Fig. 5p), but they show no correlation with any other incompatible elements. Trace element contents of Qin'an loess samples are generally comparable with or lower than those of the CLP Quaternary loess, except for higher Sr, while the paleosols have comparable or higher contents of most trace elements except for lower Sr (Figs. 5-7 and Supplementary Table 1). Ratios of Nb/Ta (c. 13) and Zr/Hf (c. 38) of all our analyzed CLP loess and paleosol samples remain constant (Fig. 5m,n).

3.4 Strontium isotope ratios

Strontium isotope data for the CLP loess and paleosols are given in Supplementary Table 2. The ⁸⁷Sr/⁸⁶Sr ratios range from 0.712474 to 0.71905, which increases with increasing ⁸⁷Rb/⁸⁶Sr ratios (Fig. 7c). The CLP Quaternary loess has generally higher ⁸⁷Sr/⁸⁶Sr ratios than those of the Qin'an loess, and paleosol shows the highest ⁸⁷Sr/⁸⁶Sr ratios (Fig. 7c-e).

4. Compositional variations of the CLP loess and paleosol

235 4.1 Inheritance of carbonate minerals

The correlations between major element contents and mineral modal abundances (Fig.4g-i) suggest that the mineralogy controls on the bulk-rock variations of major element contents in the CLP loess and paleosol. As shown in Fig. 3 and Supplementary File B, carbonate minerals are important constituents of the CLP loess, which has also been reported in other CLP sections (e.g., Jeong et al., 2008, 2011). Together with the variably high CaO and LOI and their positive correlation (Fig. 4f), the significant correlation of carbonate with CaO and LOI (Fig. 4g,h) demonstrate that calcite, not dolomite (as there is no correlation with Mg), is a significant constituent of the CLP loess and controls. Furthermore, CaO and LOI do not correlate with any other element, except for Sr. As shown in Figs. 4g&7a, it is the carbonate modes that control the abundances of CaO and Sr as well as Rb/Sr ratios.

Aluminum has the lowest water-rock partition coefficient compared with other major elements during chemical weathering and can be inherited from source rocks through the transformation of feldspar to clay minerals (such as illite and kaolinite). Thus, Al₂O₃ has been taken as the constant index and its relationship with other elements has been used to identify the mobility/immobility of these elements during sedimentary processes (e.g., Taylor et al., 1983). The positive correlations of Al_2O_3 with SiO₂-TiO₂-K₂O-FeOT-MnO (Fig. 4) indicate the controls of silicate minerals (e.g., quartz, feldspar, and phyllosilicate minerals) on the varying SiO₂-TiO₂-K₂O-

FeOT-MnO abundances. Together with the negative correlations of Al₂O₃ with CaO-LOI (e.g., Fig. 4a), it reflects the complementary modal relationship between calcite and silicate minerals in the CLP loess and paleosols (i.e., the more calcite, the less silicate minerals, and vice versa). Because carbonate minerals mainly occur as detrital grains indicated by its morphology (Fig. 3 and Supplementary File B), they are most likely derived from the provenance. Thus, the common presence of carbonate reflects the inherited signatures of high modes of carbonate and high CaO of source rocks rather than the influence by post depositional processes. Considering that both periglacial and peridesert loess are the product of an arid environment, rainfall cannot supply sufficient CaO for the precipitation of abundant carbonate in the CLP loess. Even if there were secondary carbonate minerals formed after loess deposition, they are likely re-precipitated following the dissolution of local primary carbonate, which is still of provenance origin. Hence, the common presence of calcite and high CaO-LOI in the CLP loess are most likely inherited from the provenance, which is dominated by siliciclastic and carbonate sedimentary rocks, granitoids, and their metamorphic equivalents (Jeong et

important in revealing the "true" compositional makeup of the CLP loess.

al., 2008). Thus, our bulk-rock analysis without prior chemical leaching is critically

4.2 The compositional variations

4.2.1 The compositional inheritance

275 Most of the analyzed elements, including REEs–Th, Nb–Ta, transition metals and

Ga-Sn, show significant correlations with La (Figs. 2b, 5a-k&6a-f). These

correlations reflect the inheritance of these elements from the source rocks with

limited chemical alterations during loess formation. The relatively constant Zr-Hf

content with La is probably caused by differential physical separation of zircons (and

possibly other heavy minerals; Fig. 51). As previous studies reported for various

minerals in river sediments (e.g., Garçon et al. 2014), epidote group minerals (epidote,

allanite, zoisite), titanite and clay minerals are significant for hosting REEs and Th;

zircon and garnet are important hosts for HREEs, while zircon, rutile, and titanite are

For water-soluble elements, previous studies only showed good correlations of Li

with LREEs in loess (Sauzéat et al., 2015). However, this study shows that water-

soluble elements (i.e., Cs-Rb-Ba-K) of loess are also significantly correlated with La

in this study except Li (Fig. 6). These correlations indicate the conservation of these

water-soluble elements during sedimentary processes to produce loess. Furthermore,

the correlations of total modal abundance of micas, K-feldspar and clay minerals with

Cs-Rb-Ba-K-Li (Figs. 4i&6g,h) reflect the accommodation of these elements by

micas, K-feldspar and clay minerals in the loess, the latter of which may inherit from

the weathered source rocks. Because different trace elements are preferentially hosted

in different minerals, the conservation of important mineral hosts during sedimentary

processes as single particles or rock fragments is important to preserve geochemical

hosts for HFSEs (Supplementary File B).

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signatures of the provenance.

Molybdenum and U are correlated (Fig. 5p), but do not correlate with any other

elements. Moreover, in contrast to the constant Th/Nb ratio, Th/U ratios of the CLP loess vary significantly from 2.5 to 5.3 with an average value of 4.0 (Fig. 5o). Given the insignificant chemical weathering under cold and arid conditions during sedimentary processes to produce loess, the variations of Mo and U may inherit geochemical signatures of bedrocks caused by previous weathering history. Because both Mo and U are sensitive to oxidation, this suggests that the source rocks may have suffered significant oxidation. As Carpentier et al. (2013) suggested, higher Th/U ratios of sediments derived from mature continental areas than those from juvenile terranes reflect the loss of U resulting from the long-term weathering history of source materials.

4.2.2 The effects of chemical alterations on paleosol

Although paleosol shows similar correlations between various elements like the loess, the paleosol is characterized by less carbonate with more clay minerals (Fig. 3), lower CaO-Na₂O-P₂O₅-LOI-Sr (Fig. 4) and higher ⁸⁷Sr/⁸⁶Sr ratios (Fig. 7c-e) relative to the loess.

Paleosol is thought to have experienced stronger chemical weathering and biological alterations than loess. By using the approach of McLennan (1993) to calculate values of the chemical index of alteration (CIA = $100 \times \text{molar Al}_2\text{O}_3/[\text{Al}_2\text{O}_3$ + CaO* + K₂O + Na₂O], and CaO* is corrected for apatite and carbonate), paleosol has higher CIA with an average of 64 than those of the CLP loess (i.e., 58 and 61 for average CLP Quaternary loess and Qin'an loess, respectively). The stronger weathering of the paleosol, as indicated by its higher CIA values, can result in 15/30

leaching and is responsible for subsequent reprecipitation of calcite at the base of the CLP paleosol (e.g., Jahn et al., 2001). Hence, the lower CaO-LOI-Sr contents of the paleosol reflect the loss of these elements caused by carbonate dissolution during pedogenisis, which can also result in a high Rb/Sr ratio and higher ⁸⁷Sr/⁸⁶Sr ratios (Fig. 7). In addition, the paleosol lacks correlations of Na_2O and P_2O_5 with any other elements, except scattered negative trends with Al₂O₃. Together with lower Na₂O and P_2O_5 contents in the paleosol, it indicates the loss of Na and P during post depositional weathering process, and P is also possibly taken away by vegetations.

328 4.3 Temporal variations of the CLP loess

Compared to the CLP Quaternary sections, the Qin'an loess has originated from the large arid area in the Asian interior during the Early Miocene (e.g., Guo et al., 2002). Hence, the Qin'an and Quaternary sections offer the opportunity to determine if there may be any difference between CLP Miocene and Quaternary loess in terms of their dust sources as possible responses to tectonic and climate changes.

The Qin'an loess has higher CIA values than the Quaternary loess (i.e., 61 vs. 58 on average), indicating a stronger degree of chemical weathering. It also has abundances of most elements (i.e., $SiO_2-TiO_2-Al_2O_3-Fe_2O_3$ -alkali elements-Ba-Pb-REEs-HFSEs-transition metals) comparable with or lower than those of the Quaternary loess, but have higher LOI-CaO-Sr (Figs. 4–7). This is consistent with petrographically observed higher proportions of carbonate that dilutes the silicate constituents in the Qin'an loess. The well-defined linear correlation of ⁸⁷Sr/⁸⁶Sr with

⁸⁷Rb/⁸⁶Sr also reflects a clear two-component mixing relationship for both Quaternary
Loess and the older Qin'an loess (Fig. 7c). The radiogenic ingrowth is a function of
Rb/Sr ratios, which decrease with increasing carbonate component and with the
decreasing total modal abundance of micas + clay minerals + K-feldspar. Hence,
lower ⁸⁷Sr/⁸⁶Sr ratios of the Qin'an loess compared to Quaternary loess are attributed
to more carbonate and less micas + clay minerals + K-feldspar (Fig. 7d, e).

The silicate components of the Qin'an loess also differ from those of the CLP Quaternary loess as indicated by different slopes in element covariation diagrams (Figs. 5 & 6). The significantly lower Rb relative to the difference of Sr reflects that the silicate component of Qin'an loess has lower Rb/Sr ratios and incompatible elements than Quaternary loess (Fig. 7f,g). Previous studies have also found the systematic differences of Sr-Nd-Pb-O isotopic and chemical compositions in silicate fractions after removing carbonate between those from Miocene loess and Quaternary loess (Chen and Li, 2013; Sun and Zhu, 2010; Sun et al., 2020).

Moreover, although the pseudochron ages have no chronological significance for the loess deposition, this study gives an older pseudochron age of the Qin'an loess than the Quaternary loess, i.e., 352 vs. 246 Ma (Fig. 7c). All these variations can be attributed to the change of the source material contribution, which is as the function of tectonic uplift and climate change (An et al., 2001; Chen and Li, 2013; Sun and Zhu, 2010; Sun et al., 2010, 2020).

5. Representativeness of the CLP loess about the UCC composition

362 Unlike large-scale surface sampling of shields or orogens, fine-grained
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terrigenous sediments can supply the naturally well-mixed composition of eroded surfaces, eliminating the deviation caused by biased sampling or weighted averaging that may plague surface sampling (e.g., Condie, 1993). Among fine-grained terrigenous sediments (e.g., shales), loess, especially that of peridesert origin, is an excellent candidate to estimate the average UCC composition, because of ignorable chemical weathering and a relatively long-distance transport from large source areas. It has been shown that the CLP loess deposits are the thickest and most extensive loess deposit on Earth, and the volume of dust eroded in total can be up to 190,584 km³ given a thickness of 300 m (Li et al., 2020). Moreover, as source regions of the CLP, bedrocks of surrounding deserts (Taklamakan, Gobi, Badain Jaran and Tengger) and mountains (Altyn, Qilian, Kunlun, Tianshan and Gobi Altay; Fig. 1b), covering over 2 million square kilometers, ean be dated back from Archean to Cenozoic (e.g., Nie et al., 2014). Hence, the CLP loess is better representative for estimating the average UCC composition than other loess deposits (e.g., Chauvel et al., 2014; Sauzéat et al., 2015).

To estimate the UCC composition, we used 30 CLP Quaternary loess samples in this study. For those elements that were also reported by Chauvel et al. (2014; everything except Be-Ga-Mo-Sn-Tm-W), we also combined their data for 13 CLP loess samples with the data reported here to estimate the average composition of the UCC (Table 1). We did not use paleosol samples because of the potential alteration associated with pedogenesis, such as the Sr loss in paleosol as mentioned above. The Miocene Qin'an loess is also excluded to better constrain the UCC composition

because of different correlations of various elements with La (Figs. 5-7) and stronger
effects of chemical weathering as indicated by their higher CIA values.

387 5.1 Major elements

The average contents of major elements of the CLP Quaternary loess agree well with recommended values of the average UCC composition (Rudnick and Gao, 2003) within \pm 20% variation (Fig. 8a; Table 1) except for higher CaO and lower Na₂O, the latter of which reflects the loss of Na during the sedimentary processes as discussed above. As this study and many others (e.g., Jeong et al., 2008, 2011) have shown that abundant carbonate in the CLP loess is inherited from the provenances, high CaO contents of the CLP loess reflect the well-mixed composition of exposed surfaces upwind to the west and northwest of the CLP. Gao et al. (1998) also identified higher CaO contents and more abundant carbonate in their estimates based on large-scale sampling of eastern China (up to c. 8 wt.% for the interior of the North China craton) than those estimates using exposed rocks from shields (Gao et al., 1998; Rudnick and Gao, 2003). Moreover, the loess from Kaiserstuhl, Rhine Valley has much higher CaO contents, up to c. 23 wt.%, which reflects the presence of abundant limestone from the Alps (Hu and Gao, 2008; Taylor et al., 1983). Hence, loess compositions of the CLP and Kaiserstuhl, together with large-scale exposed bedrocks in eastern China, may represent a carbonate-bearing model of the UCC composition.

404 Previous studies suggested that loess had much higher SiO₂ than the model UCC
405 composition by using the approach of large-scale surface sampling (e.g., Gallet et al.,

1998; Taylor et al., 1983) because of preferential transport of quartz into loess and its preservation during sedimentary recycling processes (e.g., Taylor et al., 1983; Rudnick and Gao, 2003). However, our study demonstrates that averaged contents of most major elements in carbonate-rich CLP loess are comparable with the proposed UCC composition including SiO_2 (Fig. 8a). As the previous study discussed (Liang et al., 2009), removing the present carbonate in the CLP loess can lead to a $\sim 3\%$ difference for Al₂O₃ content between the carbonate-leached CLP loess and those without leaching. Hence, the presence of carbonate dilutes the contents of most major elements in the CLP loess.

5.2 Trace elements

Although loess is accepted to have experienced limited chemical alteration, most water-soluble elements are known to poorly correlate with La in loess, which may reflect the heterogeneous composition of source rocks after an ineffective mixing (Fig. 9a). Hence, previous studies indirectly used surface composites or used element ratios to estimate contents of these elements in the UCC (e.g., Rudnick and Gao, 2003 and references therein). However, we find significant correlations of Li-Be-K-Rb-Cs-Ba-Pb with La and Al_2O_3 in CLP loess (Figs. 4d&6), demonstrating that these elements are congruently transported from source rocks to loess, which allow us to update recommended values of these elements in the model UCC composition.

425 Given the correlations of different elements with La (Figs. 5-6) and the widely 426 accepted La content of 31 ppm in the UCC (Rudnick and Gao, 2003), we thus use the

427	average La/X ratios (McLennan, 2001; $X = $ contents of the element of interest) to
428	determine X as the value in the UCC (Table 1). The estimates for most trace elements
429	in the mean UCC composition agree well with recommended values of the average
430	UCC composition (Rudnick and Gao, 2003) within \pm 20% variation (Fig. 8b; Table
431	1). Several elements not within uncertainty of the recommended value of Rudnick and
432	Gao (2003, those in bracket) are as follows: $Li = 34 \pm 5$ ppm (vs. 24 ppm), $Cs = 6.7 \pm 100$
433	1.2 ppm (vs. 4.9 ppm), $Ba = 461 \pm 51$ ppm (vs. 628 ppm), $Sc = 11 \pm 1$ ppm (vs. 14
434	ppm), $V = 75 \pm 11$ ppm (vs. 97 ppm), $Cr = 64 \pm 7$ ppm (vs. 92 ppm), $Co = 11 \pm 2$
435	ppm (vs. 17 ppm), $Ni = 29 \pm 4$ ppm (vs. 47 ppm), $Cu = 21 \pm 4$ ppm (vs. 28 ppm), and
436	$Ga = 14 \pm 2 \text{ ppm} \text{ (vs. 17.5 ppm)}.$

Although the *Li* value in this study is higher than recommended values (18 - 24)ppm) of Rudnick and Gao (2003 and references therein), it is identical to the values derived in recent studies, i.e., 35 ± 11 ppm based on correlations between Li and Nb in shales using MC-ICP-MS (Teng et al., 2004), 31 ± 5.2 ppm based on Neoproterozoic and Phanerozoic glacial diamictites (Gaschnig et al.2016) and comparable with the value of 30.5 ± 3.6 ppm using correlations of Li with REEs in the recent loess study by Sauzéat et al. (2015). Based on the correlation of Li with In in well-characterized UCC samples (shales, pelites, loess, graywackes, granitoids and their composites), Hu and Gao (2008) also provided a higher recommended Li value of 41 ppm.

447 Our *Cs* value is also higher than most previous estimates (Fig. 9b), except for the
448 value of 6 ppm estimated by using sedimentary rocks (McDonough et al., 1992) and

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449	7.3 ppm for marine sediments (Plank and Langmuir, 1998). However, this value is
450	comparable with that of loess from the CLP and Tadjikistan (Supplementary Table 3;
451	Chauvel et al., 2014; Hu and Gao, 2008). Of the previous studies, Cs concentrations
452	of shield composites were only given by Gao et al. (1998; 3.6 ppm). However,
453	because of the absence of significant correlations between Cs and La in sedimentary
454	rocks, previous estimates on Cs primarily relied on the correlation of Cs with Rb and
455	used the average Rb/Cs ratio, which is calculated assuming a concentration of Rb.
456	The proposed values of Rb can range from 82 to 112 ppm in previous estimates (Hu
457	and Gao, 2008; Rudnick and Gao, 2003 and references therein), and Rb/Cs ratios used
458	also vary from 15.3 (Plank and Langmuir, 1998) to 30 (Taylor and McLennan, 1985),
459	leading to a highly variable Cs estimate (Rudnick and Gao, 2003). Therefore, this
460	study, based on the significant correlation of Cs with La, potentially yields a more
461	reliable value for Cs in the UCC.

Our **Ba** value is lower than previous estimates (Table 1 and Fig. 9b). Because of 462 the significant correlations between Ba-Rb-Cs-K and their correlations with La (Fig. 463 6), we suggest that Ba, like the alkali elements, is greatly conserved during 464 sedimentary processes. Hence, given the comparable or considerably higher Rb and 465 Cs contents of the CLP loess relative to their recommended values in the UCC, the 466 lower Ba content should reflect the source rock composition. Based on the loess data 467 of Chauvel et al. (2014), McLennan (2001), and Taylor et al. (1983), the Ba content is 468 highly variable among different loess deposits (c. 190 ppm in Kaiserstuhl to c. 810 469 ppm in Iowa; Supplementary Table 3). However, because the CLP and Tadjikistan 470

471loess as important peridesert loess deposits originated from more expansive surface472bedrocks (e.g., Chauvel et al., 2014), the consistently lower Ba contents of the loess473from these two deposits may be more representative for Ba in the UCC.474The values of Sc-V-Cr-Co-Ni-Cu are lower than the values recommended by475Rudnick and Gao (2003) and Taylor and McLennan (1985, 1995), but are generally476consistent with those of carbonate-bearing composite by Gao et al. (1998) and those

of Gaschnig et al. (2016), which are also comparable with the values of Hu and Gao (2008), except for obviously higher V (see Table 1). The values of Sc, V, and Co are also similar to those of the Canadian Shield (Eade and Fahrig, 1973; Shaw et al., 1967, 1976; Table 1). Transition metals are thought to be hosted in mafic minerals, which are preferentially incorporated into the fine-grained sediments (Garcon et al., 2014). However, the significant correlations of these elements with La support that the lower contents of these elements are inherited from their source rocks. Our Ga value is comparable with the reported Ga contents of global loess (except Kaiserstuhl with a lower Ga of c. 7 ppm; Hu and Gao, 2008) and with the recommended value of 14 ppm in the UCC by Wedepohl (1995), which adopted the value of Shaw et al. (1967) but is much lower than most other recommended values (Gao et al., 1998; Gaschnig et al., 2016; Hu and Gao, 2008; Rudnick and Gao, 2003; Taylor and McLennan, 1985, 1995).

6. Conclusions

This study presents chemical compositional data for the CLP loess, which is
 originated from expansive source regions with long geological histories involving
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multiple sedimentary cycling, long-distance transport, and effective mixing. Given the high CaO and presence of carbonate in UCC rocks of both vast western China (the primary source for the CLP) and eastern China, we propose that the CLP composition represents a possibly improved model for the carbonate-bearing UCC. This study also shows significant correlations for most elements (even water-soluble elements) with La and Al_2O_3 , indicating that most elements are conserved during the sedimentary processes that produce the CLP loess. Therefore, the ratios of La/X can provide better constraints on element values of the average UCC. The results show higher Cs (Cs = $\frac{1}{2}$) 6.7 ± 1.2 ppm), lower transition metals, Ba, and Ga. These updates on the element abundances of the CLP loess represent a possibly improved model for the carbonate-bearing UCC.

Declaration of interests

505 The authors declare that they have no known competing financial interests or 506 personal relationships that could have appeared to influence the work reported in this 507 paper.

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 - **Figure and Table captions**

Fig. 1 (a) Global distributions of loess and desert (revised after Li et al., 2020 and Chauvel et al., 2014). (b) Distribution of potential source regions for the Chinese Loess Plateau (CLP) with highlights of sampling locations in this study (b; revised after Nie et al., 2014; Sun et al., 2020). CAO – the Central Asia Orogen, NTP – the Northern Tibetan Plateau, OP – the Ordos Plateau. (c-e) Photos of loess outcrops for representative sampling localities.

Fig. 2 (a) Correlation diagram of La and Al₂O₃ contents for global loess and paleosol, compared with the recommended UCC composition (black thick cross; Rudnick and Gao, 2003). (b) Correlation diagram of La and Th contents for global loess (following Gallet et al., 1998). (c) Chondrite normalized REE patterns and (d) primitive mantle normalized trace element patterns of the CLP samples using normalization values of Sun and McDonough (1989). Model UCC compositions by Taylor and McLennan (1985, 1995) and Rudnick and Gao (2003) are plotted for comparison. The plotted

data for global loess and paleosol in the literature are compiled in SupplementaryTable 3.

Fig. 3 Backscattered electron images and corresponding identification results of
different mineral phases with their counting results by using SEM-EDS with AMICS
for representative loess and paleosol samples from CLP.

Fig. 4 Correlation diagrams of major elements and LOI (a-f) and their correlations with mineral modal abundances (g-i) for loess and paleosol samples from the CLP. Major element contents shown in (a-e) have been recalculated following a sum of major elements to be 100% for comparison with the recommended UCC composition of Rudnick and Gao (2003), as shown by thick crosses (also presented in Figs. 5-7). Literature data (Supplementary Table 3) for the CLP loess and paleosol and global loess are also plotted for comparison (also in Figs. 5&6). The symbols are the same in the following Figs. 5-7.

Fig. 5 (a-l) Correlation diagrams of representative analyzed elements (HFSEstransition metals-Ga-Sn) with La. (m-n) Correlation diagrams of Nb-Ta and Zr-Hf. (op) Co-variation diagrams between Th/Pb vs. U/Pb, and Mo and U for loess and paleosol samples from the CLP. The correlation coefficients in (a, b, d, e, g, k) also include the data on the CLP loess by Chauvel et al. (2014).

Fig. 6 Correlation diagrams of alkali and alkaline earth elements with La (a-f) and their correlations with micas + clay minerals + K-feldspar, which are thought to the important mineral hosts for these elements (g-h) for loess and paleosol samples from the CLP. The correlation coefficients are given for analytical data of the CLP loess samples of this study and updated data on the CLP loess by Chauvel et al. (2014), except Be, contents of which are not available in the latter.

Fig. 7 Correlation diagrams of Rb/Sr and ⁸⁷Sr/⁸⁶Sr ratios with CaO, Sr, Rb and
mineral modal abundances to illustrate the influence of carbonate on these
geochemical variations. More carbonate leads to increasing Sr content, while the
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decrease of micas + clay minerals + K-feldspar can lead to decreasing Rb content,
both of which will result in the decrease of Rb/Sr and ⁸⁷Sr/⁸⁶Sr ratios finally.

Fig. 8 The recommended values of the UCC composition in this study normalized to those recommended by Rudnick and Gao (2003). Major elements are in wt.% (a), while trace elements are in ppm (b). The lighter grey area represents the agreement of the recommended values within \pm 20%. The recommended values of Taylor and McLennan (1985) with updates by McLennan (2001), Gao et al. (1998; carbonatebearing composition), and Gaschnig et al. (2016) are also given for comparison.

Fig. 9 (a) The modelling diagram to show the difference between peridesert loess and periglacial loess. Although chemical alteration during the processes to produce loess is limited, Ba – Rb – Cs – K as fluid-soluble elements lack correlation with La in previously studied loess, which likely reflect the heterogeneous composition of source rocks after an ineffective mixing. Hence, the observed correlation of Ba – Rb – Cs – K with La in Fig. 6 of this study indicates that dust materials for the CLP loess has experienced effective mixing after a long-distance transport. (b) The comparison of Ba – Cs contents of this study with previous estimates, i.e., G98 - Gao et al., 1998 (including both carbonate-bearing and carbonate excluded, the plot with slash); G16 – Gaschnig et al., 2016; W95 - Wedepohl, 1995; RG03 - Rudnick and Gao, 2003; TM85 - Taylor and McLennan, 1985 (related data are given in Table 1).

Table 1: Recommended values of the UCC composition in different estimates (majorelements are in wt.%, and trace elements are in ppm).

680 Supplementary Materials

681 Supplementary Table 1: Analyzed results of loess and paleosol samples from the
682 Chinese Loess Plateau in this study (major elements are in wt.%, and trace elements
683 are in ppm).

684 Supplementary Table 2: Strontium isotope composition of loess and paleosol

685 samples from the CLP.

Supplementary Table 3: Compiled geochemical data for global loess and paleosol
samples in the literature (major elements in wt.%, trace elements in ppm).

688 Supplementary File A: Data quality of major and trace element analyses.

Supplementary File B: Petrographic studies, presenting correlations of major 690 element contents with mineral modal abundances using data in literature and 691 photomicrographs.

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🔲 Loess 🛄 Desert

}op

MG - Mongolian Gobi deserts CAO

GUR - Gurbantunggut

TG - Tengger Desert

Sampling locations:

YA - Yan'an section

HK - Hukou section

LT - Lintai section

LTN - Lantian section

Winter monsoon

Westerly wind

QA

HK

ITN

HOB - Hobq Desert MU - Mu Us sandy land

LC - Luochuan section

JZT - Jiuzhoutai section QA - Qin'an section

TK - Taklamakan Desert BJ - Badain Juran Desert QB - Qaidam Basin



Figure 2 La-Th for global loess compared with UCC

175x111mm (300 x 300 DPI)

Fig. 3 Photomicrographic images

Xiao et al., Fig.4 Co-variations of major elements

174x150mm (300 x 300 DPI)

60

Xiao et al., Fig. 5 Co-variations of transition metals

197x162mm (300 x 300 DPI)

Xiao et al., Fig. 6 Co-variations of alkali and alkaline elements

204x97mm (300 x 300 DPI)

103x126mm (300 x 300 DPI)

Figure 8 Comparison with recommended values of UCC

195x63mm (300 x 300 DPI)

Xiao et al., Fig. 9 Model diagram

Fig. 9 Modal diagram

171x50mm (300 x 300 DPI)

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1	Table 1: Recomme	nded values c	of the UCC com	position in diffe	erent estimates (1	major elements	are in wt.%, and	trace elements are in	ppm).	
									FF 2.	

No.	1	2	3	4		5	6	7		8		9		10		11	
Ref.	Shaw et al., 1967, 1976	Wedepohl, 1995ª	Fahrig and Eade, 1968; Eade & Fahrig 1973	Condie, 1993	Gao et al., 1998	Gao et al., 1998	Taylor and McLennan, 1985, 1995; McLennan, 2001	Rudnie Gao,	Rudnick and Gao, 2003		d Hu & Gao, 2008		Gaschnig et al., 2016		Others		study ^b
				R	Carbonate- free based	Carbonate- bearing	,	Value	1 sigma	Value	1 sigma	No carbonate- rich diamictite	All samples	Value	Ref.	Value	1 sigma
SiO ₂	66.8		66.2	67	67.97	64.56	65.89	66.6	1.18			70.4	70.2			66.37	1.84
TiO ₂	0.54		0.54	0.56	0.67	0.63	0.50	0.64	0.08			0.70	0.70			0.71	0.05
Al_2O_3	15.05		16.1	15.14	14.17	13.39	15.17	15.4	0.75			14.6	14.4			12.98	0.97
FeOT	4.09		4.4	4.76	5.33	5.08	4.49	5.04	0.53			4.95	5.04			4.46	0.47
MnO	0.07		0.08		0.1	0.10	0.07	0.10	0.01			0.10	0.10			0.09	0.01
MgO	2.3		2.2	2.45	2.62	2.56	2.2	2.48	0.35			2.00	2.20			2.70	0.39
CaO	4.24		3.4	3.64	3.44	8.18	4.19	3.59	0.2			1.40	1.70			7.96	2.31
Na ₂ O	3.56		3.9	3.55	2.86	2.84	3.89	3.27	0.48			1.60	1.50			1.94	0.29
K_2O	3.19		2.91	2.76	2.68	2.51	3.39	2.80	0.23			3.60	3.50			2.64	0.18
P_2O_5	0.15		0.16	0.12	0.16	0.15	0.20	0.15	0.02			0.20	0.20			0.15	0.02
Li	22				20	18	20	24	5	41	6	31.3	31.4	35 ± 11	Teng et al., 2004	34	5
Be	1.3	3.1			1.95	1.78	3	2.1	0.9	1.9	0.2	2.34	2.19			1.8	0.3
Sc	7		12	13.4	15	13	13.6	14	0.9	14	1	12.1	12			11 ^c	1
V	53		59	86	98	87	107	97	11	106	7	73.8	75			75	11
Cr	35		76	112	80	71	85	92	17	73	10	53	55.7			64	7
Со	12			18	17	16	17	17.3	0.6	15	1	11.8	12.3			11	2
Ni	19		19	60	38	35	44	47	11	34	5	26.3	27			29	4
Cu	14		26		32	29	25	28	4	27	2	20.2	20.5			21	4
Zn	52		60		70	63	71	67	6	75	9	73	72			63	9
Ga	14				18	16	17	17.5	0.7	18.6	0.8	18.2	17.6			14	2
Rb	110	110	85	83	82	71	112	82	17	94	7	110	106			92	12
Sr	316		380	289	266	273	350	320	46			91.5	99.1			259 ^d	80
Y	21		21	24	17.4	15.5	22	21	2			33.2	31			25	3
Zr	237		190	160	188	163	190	193	28			220	205			217	29

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1		2	3	4		5	6	7		8		9		10		1	1
Nb	26			9.8	12	10	12	12	1	11.6	0.5	12.7	12.4	13.7	Plank & Langmuir, 1998	12.1	1.4
Mo		1.4			0.78	0.64	1.5	1.1	0.3	0.6	0.3	0.3	0.35	1.2	Sims et al., 1990	0.8 ^d	0.3
Sn		2.5			1.73	1.51	5.5	2.1	0.5	2.2	0.2	2.41	2.38			2.4	0.3
Cs		5.8			3.55	3.02	4.6	4.9	1.5	4.9	0.3	4.3	4.3	7.3	Plank & Langmuir, 1998	6.7	1.2
Ba	1070	668	730	633	678	613	550	628	83			731	696			461	51
La	32.3		71	28.4	34.8	31	30	31	3			36.5	34.9			31°	4
Ce	65.6			57.5	66.4	58.7	64	63	4			72.9	70			63	8
Pr		6.3					7.1	7.1				8.54	8.25			7.2	0.9
Nd	25.9			25.6	30.4	26.1	26	27	2			33.3	31.5			27	3
Sm	4.61	4.7		4.59	5.09	4.45	4.5	4.7	0.3			6.84	6.36			5.3	0.6
Eu	0.937	0.95		1.05	1.21	1.08	0.88	1	0.1			1.37	1.31			1.03	0.12
Gd		2.8		4.21			3.8	4	0.3			6.05	5.6			4.8	0.6
Tb	0.481			0.66	0.82	0.69	0.64	0.7	0.1			0.983	0.92			0.73	0.09
Dy	2.9						3.5	3.9				5.57	5.28			4.2	0.5
Но	0.62						0.8	0.83				1.13	1.08			0.87	0.10
Er							2.3	2.3		2.30		3.27	3.08			2.5	0.3
Tm							0.33	0.3		0.37	0.01	0.506	0.487			0.38	0.05
Yb	1.47			1.91	2.26	1.91	2.2	2	0.4	2.34	0.03	3.07	3.02			2.4	0.3
Lu	0.233			0.32	0.35	0.3	0.32	0.31	0.05	0.36	0.01	0.479	0.464			0.37	0.05
Hf	5.8		4.3	5.12		4.45	5.8	5.3	0.7			6.5	5.98			5.5	0.7
Та	5.7	1.5		0.79	0.74	0.68	1	0.9	0.1	0.92	0.03	0.94	0.901	0.96	Plank & Langmuir, 1998	0.92	0.11
Pb		17	18	17	18	17	17	17	0.5			15.8	15.1			18	2
Th	10.3		10.8	8.6	8.95	7.71	10.7	10.5	1			11.3	10.5			11.0	1.4
U	2.45		1.5	2.2	1.55	1.39	2.8	2.7	0.6	2.6	0.1	2.66	2.4			2.7 ^d	0.5

a: Updates of Shaw et al., 1967, 1976. **b:** Values of major elements are the average composition of the CLP Quaternary losss samples recalculated to a major-element sum of 100% in this study. Considering the significant correlations of most analyzed elements with La, values of trace elements in the UCC are calculated by using the average La/X ratios (X is contents of the element of interest) and the widely accepted La content of 31 ppm. **c:** The values in red are the suggested updates of the UCC composition in this study. **d:** As Sr negatively correlates with La, we use the correlation of Sr with 1/La to estimate the Sr value of the UCC by assuming La of 31 ppm. Values of Mo and U are their average contents in the CLP loss samples, which are presented only for reference considering their differentiation from other elements during the sedimentary processes to produce the CLP loss. **e:** The recommended value is refered to Rudnick and Gao (2003).

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Supplementary File A Data quality of major and trace element analysis as well as iron isotope composition.

Table DR1. Major element analysis of international standards

			BHY	VO-2				AGV-2				GSP-2			
_	Rec. (wt.%)	Meas.	(wt.%)	AVG (wt.)	RSD (%)	RE (%)	Rec. (wt.%)	Meas. (wt.%)	RE (%)	Ref. (wt.%)	Meas.	(wt.%)	AVG (wt.)	RSD (%)	
SiO ₂	49.60	50.78	50.64	50.71	0.2	2.2	59.14	60.09	1.6	62.7 - 67.7	68.66	67.11	67.88	1.6	
TiO ₂	2.73	2.76	2.77	2.76	0.2	1.2	1.05	1.04	-1.5	0.59 - 0.92	0.68	0.66	0.67	1.9	
Al_2O_3	13.44	13.16	13.19	13.18	0.1	-2.0	17.03	16.39	-3.7	14.59 - 21.5	14.62	14.23	14.43	1.9	
Fe_2O_3T	12.39	12.15	12.16	12.15	0.1	-1.9	6.78	6.65	-2.0	4.08 - 4.96	4.96	4.85	4.91	1.5	
MnO	0.17	0.17	0.17	0.17	0.01	0.1	0.10	0.10	-0.8	0.038 - 0.06	0.0424	0.0419	0.042	0.8	
MgO	7.26	7.31	7.32	7.32	0.05	0.8	1.80	1.82	0.9	0.91 - 1.3	0.99	0.97	0.98	1.3	
CaO	11.40	11.33	11.39	11.36	0.3	-0.3	5.15	5.20	1.0	2.02 - 2.84	2.16	2.17	2.17	0.6	
Na ₂ O	2.22	2.23	2.21	2.22	0.6	0.1	4.02	4.22	5.0	1.02 - 3.62	2.86	2.84	2.85	0.5	
K ₂ O	0.51	0.53	0.51	0.52	2.3	1.0	2.90	2.93	1.2	2.89 - 5.67	5.54	5.50	5.52	0.4	
P_2O_5	0.27	0.27	0.27	0.27	1.5	0.5	0.48	0.48	-0.9	0.24 - 0.33	0.30	0.30	0.30	0.2	

Meas.: Measured values. Rec.: recommended values. AVG: average. RE: relative error between measured and recommended values. RSD: relative standard deviation. Recommended values and reference values are from <u>http://georem.mpch-mainz.gwdg.de/sample_query_pref.asp</u>.

Table DR2. Replicate analysis of major elements 16HK-03A 16HK-05B

	1	6HK-03A		1	6HK-05B	
	16HK-03A	16HK-03A	RSD	16HK-05B	16HK-05B	RSD
	(wt.%)	(wt.%)	(%)	(wt.%)	(wt.%)	(%)
SiO_2	61.47	62.12	0.7	60.30	61.13	1.0
${\rm TiO}_2$	0.65	0.65	0.4	0.62	0.62	0.2
Al_2O_3	11.73	11.83	0.6	11.20	11.25	0.3
Fe_2O_3T	4.49	4.52	0.5	4.20	4.22	0.4
MnO	0.09	0.09	2.9	0.08	0.08	0.09
MgO	2.59	2.61	0.6	2.40	2.41	0.4
CaO	7.49	7.41	0.8	8.39	8.30	0.8
Na ₂ O	2.03	2.05	0.7	1.74	1.74	0.2
K_2O	2.38	2.42	1.1	2.28	2.31	0.9
P_2O_5	0.15	0.15	0.4	0.14	0.13	3.0

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Table DR3. Trace element analysis of international standards

			Ben	-			AUV	-2		,	JSP-2
	Iso.	Ň	Jochum	et al., 2016	RE		Jochum et	al., 2016			D
		Meas.	Rec.	Uncertain ty	(%)	Meas.	Rec	Uncert ainty	- KE (%)	Meas.	Reī.
Li	7	8.76	9.13	0.22	-4.0	10.48	10.8	0.21	-3.0	36.12	31.1 - 5
Be	9	2.33	2.17	0.1	7.5	2.11	2.209	0.066	-4.4	1.49	0.92 - 1
Sc	45	32.15	33.53	0.4	-4.1	12.22	13.11	0.31	-6.8	6.56	5 - 8.3
V	51	435	417.6	4.5	4.1	122	118.5	1.2	2.9	54.03	42 - 5
Cr	53	14.19	15.85	0.38	-10	15.20	16.22	0.72	-6.3	18.27	15.6 - 6
Co	59	36.55	37.33	0.37	-2.1	15.15	15.46	0.5	-2.0	6.98	3.72 -
Ni	60	11.06	12.57	0.3	-12	17.39	18.87	0.41	-7.9	14.83	1.99 -
Cu	63	17.29	19.66	0.72	-12	49.11	51.51	0.65	-4.6	38.75	22.6 -
Zn	66	133	129.5	1.8	3.1	90.67	86.7	1.2	4.6	120	75 - 13
Ga	71	21.14	22.07	0.19	-4.2	19.60	20.42	0.17	-4.0	22.84	19.6 - 2
Rb	85	45.71	46.02	0.56	-0.7	65.85	67.79	0.66	-2.9	239	193 - 2
Sr	88	344	337.4	6.7	1.8	668	659.5	5.7	1.3	242	209 - 2
Y	89	33.61	36.07	0.37	-6.8	18.44	19.14	0.84	-3.7	25.49	22 - 3
Zr	90	181.9	186.5	1.5	-2.5	230	232	2.3	-1.0	550	139 - (
Nb	93	11.92	12.44	0.2	-4.2	13.61	14.12	0.22	-3.6	26.52	18 - 3
Mo	95	248	250.6	6.7	-0.9	1.87	2	0.11	-6.4	2.21	1.34
Sn	118	1.94	2.28	0.13	-15	1.82	1.83	0.25	-0.4	6.54	1.184-8
Cs	133	1.13	1.16	0.023	-2.5	1.16	1.173	0.018	-1.0	1.19	0.98 -
Ba	137	694	683.9	4.7	1.5	1172	1134	8	3.4	1411	1149 -
La	139	24.87	25.08	0.16	-0.9	38.18	38.21	0.38	-0.08	192	159 - 2
Ce	140	52.50	53.12	0.33	-1.2	69.77	69.43	0.57	0.5	457	343.9 -
Pr	141	6.56	6.83	0.044	-3.9	8.07	8.165	0.084	-1.1	58.23	47.1 - 0
Nd	146	27.84	28.26	0.37	-1.5	29.87	30.49	0.47	-2.0	213	177.8 -
Sm	147	6.37	6.55	0.047	-2.7	5.43	5.509	0.078	-1.4	26.01	23.1 - 2
Eu	151	1.91	1.99	0.024	-3.8	1.54	1.553	0.015	-1.0	2.24	1.96 - 2
Gd	156	6.69	6.81	0.078	-1.8	4.43	4.678	0.064	-5.3	12.87	10.1 -
Tb	159	1.02	1.08	0.026	-5.0	0.63	0.651	0.0073	-3.6	1.31	1.09-1
Dy	163	6.04	6.42	0.055	-5.9	3.32	3.549	0.031	-6.4	5.50	5.08 -
Ho	165	1.26	1.31	0.011	-3.8	0.66	0.68	0.0081	-2.7	0.95	0.8 - 1
Er	167	3.50	3.67	0.038	-4.6	1.78	1.825	0.013	-2.6	2.25	2.02 - 2
Tm	169	0.52	0.53	0.006	-2.9	0.26	0.2623	0.0035	-1.9	0.30	0.23 -
Yb	173	3.28	3.39	0.036	-3.4	1.55	1.653	0.013	-6.2	1.62	1.23 -
Lu	175	0.49	0.50	0.0078	-2.5	0.25	0.2507	0.0033	0.02	0.23	0.16 -
Hf	178	4.81	4.97	0.034	-3.2	5.08	5.137	0.057	-1.0	13.71	1.81 -
Та	181	0.81	0.78	0.018	2.7	0.87	0.865	0.019	0.04	0.88	0.43 -
W	182	0.47	0.46	0.05	1.2	0.50	0.553	0.094	-9.9	0.35	0.29 -
Pb	208	9.66	10.59	0.17	-8.8	12.70	13.14	0.15	-3.3	41.65	35.38
Th	232	5.81	5.83	0.05	-0.3	6 14	6 174	0.063	-0.6	113	92.20
111	232	5.01	5.05	0.05	0.5	0.14	0.174	0.005	0.0	115	10

Meas.: Measured values in ppm. Iso.: isotope. Rec.: recommended values in ppm. Ref.: reference values in ppm. RE: relative error between measured and recommended values. RSD: relative standard deviation. Recommended values and reference values are from http://georem.mpch-mainz.gwdg.de/sample_query_pref.asp.

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Table DR4. Replicate analysis of trace elements

	1	6LTN-02a			16QA-04a			16LC-06a			JZT16-05a	
	200528A 41.d	200528 A51.d	RSD (%)	200528 A34.d	200528 A55.d	RSD (%)	200528 A19.d	200528 A54.d	RSD (%)	200528 A26.d	200528 A53.d	RSD (%)
Li	40.93	42.81	3.2	40.72	42.76	3.5	40.98	42.08	1.9	34.38	34.27	0.2
Be	2.17	2.28	3.3	2.07	2.01	1.9	2.28	2.32	1.4	1.73	1.86	5.0
Sc	12.64	12.49	0.8	11.10	11.23	0.8	13.59	13.52	0.4	10.58	10.39	1.3
V	94.66	95.40	0.5	82.01	83.01	0.9	105.8	107.0	0.8	75.58	76.70	1.0
Cr	72.56	72.86	0.3	66.41	66.80	0.4	82.19	79.99	1.9	68.17	68.86	0.7
Co	13.61	13.68	0.4	11.63	11.99	2.1	14.95	15.27	1.5	10.31	10.49	1.2
Ni	35.57	35.38	0.4	31.35	31.83	1.	41.47	41.91	0.7	28.14	28.51	0.9
Cu	24.98	24.98	0.0	21.22	21.40	0.6	29.11	29.51	1.1	19.96	20.11	0.5
Zn	72.20	71.86	0.3	66.33	65.53	0.9	85.09	81.67	2.9	59.05	60.43	1.6
Ga	16.09	16.07	0.1	13.96	14.28	1.6	17.18	17.32	0.5	13.02	13.19	0.9
Rb	105.6	104.9	0.4	98.16	100.8	1.9	116.9	118.0	0.6	90.51	91.03	0.4
Sr	155.3	154.4	0.4	320.8	326.2	1.2	144.8	146.6	0.8	419.5	430.1	1.7
Y	27.24	26.99	0.7	25.40	26.30	2.5	28.16	28.26	0.3	27.17	25.13	5.5
Zr	214.2	203.5	3.6	205.4	214.7	3.1	248.4	233.3	4.4	285.7	256.7	7. (
Nb	13.90	13.70	1.0	12.60	12.80	1.1	14.77	16.78	9.0	11.91	11.94	0.1
Мо	0.87	0.84	2.6	0.87	0.90	2.1	0.91	0.93	1.2	1.03	1.11	5.1
Sn	2.75	2.77	0.4	2.56	2.59	0.7	2.98	2.89	2.4	2.36	2.35	0.5
Cs	8.62	8.63	0.03	7.91	8.05	1.3	9.47	9.57	0.8	6.11	6.26	1.7
Ba	533.5	529.9	0.5	501.0	508.0	1.0	545.5	550.8	0.7	503.1	505.6	0.3
La	36.67	36.54	0.2	33.34	32.02	2.9	37.82	36.89	1.7	32.46	31.94	1.1
Ce	73.71	73.72	0.02	67.15	64.42	2.9	81.68	79.41	2.0	66.00	64.54	1.6
Pr	8.41	8.38	0.3	7.64	7.33	3.0	8.82	8.63	1.5	7.60	7.34	2.4
Nd	30.89	30.35	1.2	28.76	27.61	2.9	32.32	31.71	1.3	28.21	26.86	3.5
Sm	6.14	6.13	0.2	5.64	5.54	1.3	6.37	6.35	0.2	5.65	5.33	4.
Eu	1.18	1.17	0.8	1.10	1.06	2.9	1.26	1.25	0.7	1.06	1.07	0.5
Gd	5.58	5.58	0.03	5.20	5.23	0.4	5.53	5.72	2.5	5.31	5.04	3.7
Tb	0.83	0.84	1.0	0.78	0.79	1.6	0.87	0.87	0.6	0.79	0.75	3.3
Dy	4.66	4.75	1.3	4.40	4.54	2.2	4.97	5.05	1.0	4.57	4.24	5.2
Но	0.99	0.97	1.5	0.93	0.96	2.0	1.04	1.04	0.1	0.96	0.89	5.3
Er	2.77	2.77	0.07	2.59	2.70	2.9	2.97	2.89	2.0	2.82	2.53	7.4
Tm	0.42	0.43	0.07	0.40	0.42	3.4	0.46	0.46	0.5	0.46	0.40	9.2
Yb	2.69	2.67	0.5	2.54	2.60	1.7	2.98	2.85	3.0	2.97	2.51	11.
Lu	0.42	0.43	1.0	0.39	0.41	2.7	0.46	0.44	2.5	0.47	0.39	12.
Hf	5.56	5.37	2.4	5.29	5.58	3.8	6.51	6.15	4.0	7.15	6.61	5.5
Та	1.08	1.07	0.8	1.01	1.03	1.5	1.14	1.29	8.8	0.92	0.93	0.9
W	2.26	2.26	0.1	2.02	2.03	0.1	2.41	2.46	1.3	1.82	1.82	0.2
Pb	21.68	21.53	0.5	19.42	19.62	0.7	23.95	24.09	0.4	18.09	18.11	0.0
Th	13.27	13.14	0.7	12.02	12.05	0.2	14.44	14.17	1.3	12.26	12.11	0.8
IJ	2 49	2 49	0.01	3 16	3 18	0.6	2 86	2 76	25	4 55	4 39	24

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Supplementary File B Petrographic studies, presenting co-variations of major element contents with mineral modal abundances using data in literature and photomicrographs of our samples.

Figure A. Co-variation diagrams of bulk-rock major element contents with mineral modal abundances for the CLP loess and paleosol (all these compiled data are referred to Jeong et al., 2008, 2010, whose data are systematic). (a) SiO_2 is significantly correlated with quartz. (b-c) Na₂O positively correlates with plagioclase while negatively correlates with phyllosilicate. (d-g) $AI_2O_3 - TFe_2O_3 - MnO - K_2O$ are correlated with phyllosilicate (i.e., micas, kaolinite, chlorite, and other clay minerals such as illite, smectite, vermiculite). (h-i) Co-variations of MgO with minerals, reflecting its correlation with chlorite. (j) CaO is strictly correlated with both calcite and phyllosilicate minerals.

Figure B. Representative petromicrographs of the CLP loess samples, showing the common presence of carbonate.

References:

- Jeong, G.Y., Hillier, S., Kemp, R.A., 2008. Quantitative bulk and single-particle mineralogy of a thick Chinese loess–paleosol section: implications for loess provenance and weathering. Quaternary Science Reviews 27, 1271-1287.
- Jeong, G.Y., Hillier, S., Kemp, R.A., 2011. Changes in mineralogy of loess–paleosol sections across the Chinese Loess Plateau. Quaternary Research 75, 245-255.

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				Qin'an Loess					Luochua	an Loess				Luo	chuan Pale	osol			
Locatio	n	34°56'21"N,	105°33'7"E	35°2'24"N,	34°59'N,	105°32'E					35	°42'N, 109°	25'E					36°9'N,	110°26'E
Sample	е	16QA-01a	16QA-02a	16QA-03A	16QA-04a	16QA-05a	16LC-01a	16LC-03a	16LC-05	16LC-07	16LC-09a	16LC-10a	16LC-02a	16LC-04a	16LC-06a	16LC-08a	16LC-11a	16HK-01a	16HK-02a
SiO2		47.60	38.52	53.32	55.64	57.16	58.39	59.68	62.72	62.06	64.59	62.44	62.72	64.92	62.90	61.42	65.75	59.44	64.63
TiO2		0.53	0.46	0.54	0.61	0.62	0.62	0.62	0.72	0.71	0.69	0.70	0.71	0.75	0.76	0.71	0.73	0.59	0.62
AI2O3	3	10.14	8.60	9.99	11.84	12.08	11.28	11.40	13.29	12.96	11.66	12.72	12.97	13.76	13.87	13.34	13.28	10.94	11.57
TFe2O	3	3.91	3.33	3.61	4.46	4.55	4.33	4.31	5.33	5.19	3.73	4.85	5.11	5.46	5.63	5.30	5.34	4.05	4.14
MnO		0.08	0.09	0.05	0.080	0.086	0.08	0.09	0.10	0.09	0.04	0.12	0.10	0.10	0.11	0.10	0.10	0.08	0.07
MgO		2.19	1.99	1.90	2.66	2.55	2.22	2.15	2.43	2.31	2.06	2.43	2.14	2.19	2.32	2.38	2.20	2.60	2.37
CaO		16.04	22.40	13.67	10.02	9.80	9.34	8.57	5.41	6.21	6.14	6.08	4.32	3.07	4.44	5.27	3.20	8.35	5.76
Na2O)	1.02	0.90	1.24	1.64	1.63	1.72	1.68	1.28	1.20	1.65	1.31	1.65	1.59	1.02	1.10	1.29	1.84	2.02
K2O		2.05	1.83	2.08	2.17	2.13	2.30	2.30	2.68	2.63	2.39	2.57	2.74	2.70	2.76	2.73	2.76	2.24	2.32
P2O5		0.15	0.17	0.14	0.12	0.15	0.13	0.13	0.12	0.13	0.12	0.16	0.12	0.10	0.11	0.12	0.10	0.14	0.13
LOI		16.68	21.27	15.33	11.38	11.20	10.71	9.82	7.72	8.04	8.31	9.31	6.53	6.06	7.89	8.77	6.62	9.87	7.52
TOTA	L	100.40	99.57	101.88	100.61	101.95	101.13	100.73	101.80	101.52	101.36	102.69	99.10	100.70	101.80	101.24	100.79	100.13	101.16
	-		00.0	00.4	40 7	40 7		00.0	00.0	00.4	00.0	00 7		40.4	44.0	00.0	o 4 -	05.4	04.0
	(38.8	33.2	38.1	40.7	40.7	34.2	33.2	39.0	38.1	33.6	39.7	47.4	49.4	41.0	39.9	34.7	35.1	31.2
Ве	9	1.64	1.39	1.67	2.07	1.93	1.85	1.91	2.20	2.07	1.88	2.04	2.32	2.21	2.28	2.10	1.90	1.72	1.69
50	40	10.1	0.52	9.07	020	026	70.2	0.0	13.1	12.0	76.2	12.3	12.9	101 5	105.0	10.0	12.0	10.5	75.0
V Cr	51	75.3	62.9	00.0 56.0	02.0	03.0 65.0	79.3	60.9	90.5	94.5 75.6	70.3	91.5	95.1	01.0	105.0	74.0	09.1	/0.Z	75.0
	55	00.4	20.4 9.26	50.2 7 10	00.4 11.6	11 0	03.7	01.4	14.0	12.0	10.5	11.2	12.7	01.0	02.2	14.0	16.0	00.0 10.7	03.0
Ni	59	9.07	24.0	7.19	31.3	31.6	20.8	30.1	38.5	38.0	20.1	34.3	37.1	14.0	14.9	30.7	37.0	10.7	27.4
Cu	63	29.0	24.9 18.6	20.1	21.2	22.1	29.0	20.8	26.6	27.1	10.2	26.4	24.7	40.2 27.5	20.1	27.7	26.3	20.0	10.1
Zn	66	64.0	54 1	57.9	66.3	69.2	64.6	63.2	79.3	75.7	63.0	76.8	70.5	75.5	85.1	80.3	72.9	62.6	58.8
Ga	71	12.6	10.4	11.6	14.0	14.3	13.8	13.7	16.4	15.9	14 1	15.8	16.2	17.0	17.2	16.8	16.0	13.4	13.8
Rb	85	81.2	69.8	77.1	98.2	99.5	95.3	94.3	112	10.0	95.6	102.7	114	116	117	114	105	89.7	86.7
Sr	88	293	325	236	321	299	202	200	165	154	187	175	174	183	145	156	139	313	249
Y	89	20.6	19.1	20.2	25.4	26.3	24.8	25.6	31.3	27.9	27.5	26.3	29.5	29.6	28.2	27.9	25.9	24.8	22.8
Zr	90	182	139	179	205	207	220	231	221	238	275	217	250	238	248	227	241	255	222
Nb	93	10.5	8.7	10.4	12.6	12.4	12.5	12.5	14.3	14.0	13.5	13.7	14.5	14.8	14.8	14.4	13.6	12.0	12.3
Мо	95	0.65	0.71	0.50	0.87	1.09	0.63	0.75	0.77	0.81	0.22	0.48	0.75	1.08	0.91	0.86	0.88	0.74	0.48
Sn	118	2.01	1.69	1.94	2.59	2.58	2.51	2.40	2.79	2.70	2.53	2.66	2.95	3.03	2.89	2.92	2.83	2.32	2.07
Cs	133	6.76	5.63	6.35	7.91	8.04	7.01	6.68	8.78	8.48	6.93	7.96	9.02	9.43	9.47	9.08	8.24	6.20	5.29
Ва	137	339	272	306	501	509	476	478	517	512	442	485	556	578	546	543	492	490	523
La	139	25.6	22.4	25.3	33.3	33.1	32.2	33.5	38.8	36.4	35.5	36.0	38.1	39.9	37.8	36.7	33.1	32.1	32.8
Ce	140	51.8	44.8	50.5	67.1	66.0	64.6	67.4	74.4	73.8	72.1	74.7	76.9	81.9	81.7	77.6	71.4	64.2	66.3
Pr	141	5.94	4.98	5.77	7.64	7.46	7.43	7.70	8.69	8.39	8.18	8.46	8.74	9.16	8.82	8.61	7.84	7.35	7.56
Nd	146	22.2	18.7	21.5	28.8	28.4	27.1	27.8	31.8	30.9	30.0	30.8	32.0	33.5	32.3	31.1	29.5	27.0	27.8
Sm	147	4.41	3.70	4.34	5.64	5.63	5.44	5.63	6.35	6.05	5.99	6.14	6.43	6.76	6.37	6.27	5.80	5.40	5.47
Eu	151	0.91	0.76	0.86	1.10	1.11	1.06	1.10	1.25	1.18	1.15	1.19	1.24	1.29	1.26	1.22	1.16	1.07	1.08
Gd	156	4.10	3.52	4.13	5.20	5.25	5.00	5.08	5.94	5.47	5.47	5.45	5.76	5.93	5.53	5.57	5.13	4.85	4.67
Tb	159	0.62	0.53	0.61	0.78	0.79	0.75	0.78	0.91	0.84	0.83	0.83	0.87	0.90	0.87	0.87	0.76	0.75	0.69
Dy	163	3.51	3.05	3.50	4.40	4.55	4.24	4.48	5.25	4.81	4.74	4.70	5.10	5.21	4.97	4.90	4.88	4.24	4.02
Ho	165	0.73	0.64	0.74	0.93	0.96	0.90	0.94	1.09	0.99	0.98	0.98	1.09	1.08	1.04	1.02	0.98	0.89	0.83
Er _	167	2.11	1.89	2.08	2.59	2.72	2.54	2.62	3.08	2.81	2.80	2.74	3.06	3.07	2.97	2.84	2.95	2.56	2.37
Tm	169	0.33	0.29	0.32	0.40	0.42	0.38	0.41	0.47	0.43	0.44	0.42	0.47	0.46	0.46	0.44	0.44	0.38	0.35

Yb	173	2.07	1.85	2.08	2.54	2.66	2.51	2.58	2.97	2.77	2.77	2.56	3.01	3.00	2.98	2.85	2.86	2.43	2.30
Lu	175	0.32	0.28	0.32	0.39	0.40	0.39	0.40	0.47	0.43	0.42	0.40	0.46	0.46	0.46	0.43	0.42	0.38	0.35
Hf	178	4.62	3.67	4.57	5.29	5.45	5.65	5.99	5.74	6.12	6.96	5.60	6.44	6.26	6.51	5.91	6.56	6.36	5.61
Та	181	0.81	0.66	0.79	1.01	0.96	0.97	0.99	1.08	1.09	1.08	1.03	1.12	1.13	1.14	1.18	1.04	0.91	0.93
W	182	1.55	1.26	1.48	2.02	2.08	1.98	1.91	2.49	2.24	1.90	2.18	2.34	2.45	2.41	2.40	2.39	1.81	1.79
Pb	208	17.2	14.1	16.0	19.4	19.9	18.3	18.5	22.0	22.0	18.3	20.3	22.1	23.3	24.0	23.1	21.1	17.1	16.7
Th	232	9.31	7.59	8.76	12.0	12.2	11.2	11.5	12.9	12.7	12.0	12.9	13.6	14.7	14.4	13.6	12.3	11.5	10.0
U	238	2.58	2.32	2.52	3.16	3.13	2.62	2.74	2.62	2.59	2.61	2.48	2.86	2.91	2.86	2.73	2.63	3.18	2.54

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		Yan'an-Huko	ou Loess				Lantian Loes	s	Lantian	Paleosol		Lingtai	Loess				
Location	n 36°10'N,	110°26'E	36°10'25"N, 110°25'50"E	36°37'22"N, 109°28'35"E	36°37'58"N, 109°24'28"E		34	°10'N, 109°	11'E		35°2'00"N, 107°31'6"E	34°	°58'N, 107°3	33'E			
Sample	16HK-03a	16HK-04a	16HK-05B	16YA-01a	16YA-02a	16LTN-01	16LTN-02a	16LTN-03	16LTN-04a	16LTN-05a	16LT-01a	16LT-02a	16LT-03a	16LT-04A	JZT16-01a	a JZT16-02a	JZT16-03a
SiO2	61.47	60.22	60.30	65.05	62.52	61.37	58.95	65.05	67.11	65.34	59.07	68.43	62.55	59.46	60.69	58.85	58.68
TiO2	0.65	0.63	0.62	0.67	0.60	0.72	0.68	0.76	0.77	0.75	0.63	0.77	0.71	0.67	0.61	0.60	0.60
AI2O3	11.73	11.49	11.20	12.23	10.95	13.48	12.86	14.77	14.67	15.67	11.40	14.36	13.03	12.43	10.92	11.21	11.03
TFe2O3	4.49	4.29	4.20	4.67	4.05	5.43	5.04	5.73	5.77	6.09	4.38	5.71	5.17	4.91	4.15	4.21	4.19
MnO	0.09	0.08	0.08	0.08	0.07	0.10	0.09	0.105	0.10	0.10	0.08	0.11	0.10	0.09	0.08	0.08	0.08
MgO	2.59	2.56	2.40	2.35	2.24	2.13	2.07	2.09	2.06	1.91	2.39	2.11	2.08	2.23	2.65	2.73	2.80
CaO	7.49	7.45	8.39	5.24	7.98	6.04	7.40	2.60	1.21	1.02	9.56	1.12	5.47	7.13	8.39	8.56	8.16
Na2O	2.03	2.06	1.74	1.72	1.79	1.60	1.57	1.63	1.60	1.07	1.67	1.67	1.69	1.65	1.79	2.18	2.27
K2O	2.38	2.24	2.28	2.46	2.22	2.55	2.49	2.51	2.79	2.47	2.25	2.91	2.74	2.54	2.36	2.41	2.37
P2O5	0.15	0.14	0.14	0.11	0.13	0.13	0.14	0.14	0.14	0.07	0.13	0.19	0.16	0.15	0.14	0.15	0.15
LOI	8.94	10.20	9.48	7.54	8.67	8.80	9.83	7.33	5.18	6.39	10.58	4.30	8.06	8.97	9.12	9.58	10.28
TOTAL	102.00	101.36	100.83	102.12	101.23	102.35	101.12	102.72	101.41	100.87	102.13	101.66	101.74	100.23	100.91	100.57	100.60
Li	36.8	34.5	34.5	32.9	29.1	44.5	40.9	45.5	39.6	50.5	36.7	41.7	37.4	36.8	30.4	35.2	32.8
Be	2.04	1.77	1.96	1.80	1.53	2.38	2.17	2.22	3.04	2.55	1.94	2.64	2.17	1.94	1.57	1.83	1.71
Sc	11.3	11.6	10.9	10.4	11.0	13.0	12.6	13.8	14.3	15.1	11.1	14.1	12.7	12.7	11.3	10.9	11.4
V	82.5	73.5	80.9	74.9	65.8	98.8	94.7	104.4	105.5	109.5	82.7	101.9	92.8	81.1	68.9	79.6	71.8
Cr	68.5	66.1	65.2	60.1	58.0	74.0	72.6	79.4	77.3	83.4	65.1	75.2	72.1	67.6	66.8	69.2	65.6
Co	11.7	11.2	11.0	10.3	9.9	14.0	13.6	15.0	15.3	15.8	11.4	15.0	13.7	12.6	10.3	11.2	10.9
Ni	31.1	30.2	28.8	28.0	27.4	36.9	35.6	39.8	39.6	44.8	31.0	41.3	36.8	34.6	28.8	29.5	30.2
Cu	21.2	21.2	19.8	18.3	18.5	26.7	25.0	27.1	27.9	28.1	21.0	28.9	25.3	24.8	20.2	20.8	21.3
Zn	68.4	64.8	63.2	57.2	56.2	74.8	72.2	78.4	81.4	82.2	62.8	84.1	76.5	71.8	59.2	63.6	65.4
Ga	14.4	14.2	13.6	13.0	12.9	16.7	16.1	17.8	18.0	19.1	13.9	17.7	16.2	15.4	12.9	13.7	13.8
Rb	96.2	88.7	92.2	87.7	82.3	110.4	105.5	117.5	118	117	92.1	121.3	111.1	102.0	85.0	94.7	90.2
Sr	301	264	253	229	215	152	155	142	172	138	287	134	181	167	262	292	248
Y	24.3	24.2	25.7	24.3	22.5	28.9	27.2	31.4	32.6	29.4	25.9	32.7	28.7	25.2	23.5	25.7	24.6
Zr	220	222	229	242	221	221	214	230	241	251	218	254	238	198	251	236	211
Nb	13.0	12.4	12.6	11.7	11.8	14.2	13.9	15.3	15.6	15.1	12.3	15.6	14.2	13.1	11.8	12.0	12.0
Mo	1.22	0.73	0.61	0.61	0.61	0.88	0.87	0.90	0.88	1.10	0.79	0.91	0.85	0.80	1.02	0.96	0.99
Sn	2.47	2.56	2.53	2.29	2.35	3.01	2.75	3.11	3.13	3.26	2.44	3.22	2.82	2.89	2.36	2.46	2.59
Cs	7.13	6.62	6.42	5.92	5.64	9.13	8.62	9.60	9.64	10.48	6.86	9.76	8.61	7.96	5.77	6.49	6.31
Ва	523	480	476	485	445	555	533	575	589	565	494	571	536	497	469	501	480
La	32.1	31.7	31.4	31.5	28.3	38.1	36.7	39.4	40.4	38.5	33.3	41.4	36.8	34.5	29.2	31.3	32.4
Ce	65.1	64.1	62.8	63.6	57.2	77.5	/3./	80.2	82.1	82.3	67.3	82.1	75.0	69.1	59.2	63.7	65.4
Pr	7.37	7.36	7.21	7.31	6.62	8.79	8.41	8.98	9.31	9.02	7.73	9.45	8.58	7.88	6.84	7.26	7.50
Na	27.4	27.8	26.6	27.5	25.1	32.7	30.9	33.2	34.0	32.8	28.7	34.9	31.2	30.0	25.9	27.1	28.2
Sm	5.44	5.46	5.32	5.48	4.96	6.51	6.14	0.62	6.86	0.00	5.73	6.99	6.31	5.93	5.20	5.54	5.59
EU	1.08	1.11	1.06	1.03	1.00	1.24	1.18	1.27	1.30	1.32	1.10	1.34	1.22	1.17	1.05	1.07	1.08
Gđ Th	4.84	5.03	5.03	4.98	4.69	5.70	5.58	0.02	0.15	0.01	5.41	0.42	5.07	5.39	4.80	5.U8 0.77	5.20 0.75
	0.73	U./1 1 10	0.70	0.75	0.07	0.88	0.03	0.93 5 1 0	0.90	0.91	0.80	0.97	0.07	0.70	0.00 1 22	0.77	0.75
Dy Llo	4.20	4.40	4.37	4.30	4.17	0.00	4.00	0.10 1 1 1	1 10.0	0.10 1.10	4.00	0.0U	0.00 1.05	4.00	4.33	4.44	4.02
TU Er	0.00	0.00 2.72	0.93	0.90	0.04	3.04	0.99	1.11	1.10	1.10	0.94	1.10	2.00	0.94 2.79	0.07	0.93	0.91
EI Tm	2.00	2.12	2.04	0.30	2.31	0.46	2.11 0.42	0.10	0.50	0.13	2.05	0.40	2.99	2.10	2.09	2.00	2.07
	0.00	0.40	0.41	0.35	0.57	0.40	0.42	0.40	0.00	0.40	0.42	0.43	0.40	0.70	0.55	0.40	0.40

Yb	2.43	2.51	2.60	2.48	2.33	2.96	2.69	3.03	3.21	3.08	2.69	3.17	2.91	2.64	2.48	2.60	2.53
Lu	0.38	0.39	0.40	0.39	0.36	0.45	0.42	0.48	0.50	0.48	0.41	0.49	0.43	0.39	0.37	0.40	0.40
Hf	5.68	6.12	5.97	6.37	5.94	5.77	5.56	5.93	6.16	6.50	5.61	6.49	6.14	5.53	6.83	6.12	5.83
Та	0.99	0.96	0.96	0.92	0.93	1.11	1.08	1.14	1.19	1.16	0.93	1.16	1.14	1.03	0.91	0.93	0.93
W	2.03	2.08	1.90	2.09	1.90	2.39	2.26	2.43	2.56	2.54	1.99	2.43	2.31	2.26	1.90	1.84	2.02
Pb	18.9	17.9	17.8	17.3	16.4	22.8	21.7	23.4	24.2	25.4	18.8	23.9	22.2	20.3	17.7	18.5	18.3
Th	11.4	11.0	10.9	11.4	10.0	13.8	13.3	14.0	14.3	14.9	11.7	14.3	13.2	12.5	10.6	11.6	11.7
U	3.22	3.64	2.72	2.71	2.42	2.59	2.49	2.62	2.59	3.78	3.25	2.79	2.82	2.46	3.36	3.17	3.20

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		Jiuzhout	tai Loess				
Location		36°5'N,	103°47'E				
Sample	JZT16-04a	JZT16-05a	JZT16-06a	JZT16-07a	JZT16-08a	JZT16-09a	JZT16-11a
SiO2	61.17	56.53	59.78	57.25	57.17	59.30	56.97
TiO2	0.60	0.58	0.59	0.59	0.62	0.66	0.65
AI2O3	11.00	10.65	10.64	10.90	11.29	11.97	11.69
TFe2O3	4.12	4.01	3.96	4.19	4.42	4.66	4.54
MnO	0.08	0.08	0.08	0.08	0.09	0.09	0.089
MgO	2.72	2.68	2.56	2.96	2.99	3.05	2.99
CaO	8.56	9.64	8.79	9.42	8.92	6.84	8.58
Na2O	1.86	1.74	1.82	1.91	1.72	2.19	2.14
K2O	2.31	2.31	2.35	2.32	2.40	2.50	2.17
P2O5	0.14	0.14	0.15	0.15	0.16	0.15	0.16
LOI	9.51	9.16	9.17	9.73	10.07	9.34	12.84
TOTAL	102.07	97.52	99.89	99.48	99.85	100.75	102.81
	00 7	24.4	24.0	07.0	25.0	20.0	00.4
LI	30.7	34.4	31.2	37.3	35.8	38.8	38.1
Ве	1.53	1.73	1.03	2.13	1.84	1.90	1.94
Sc	10.9	10.6	11.0	70.5	74.0	11.8	11.4
V Cr	60.0	75.0	69.0 65.6	70.0 65.7	74.0	07.0	04.0 70.4
Ci	10.0	10.2	10.2	10.0	09.9 11 5	12.1	11.0
CU Ni	20.1	10.5	20.0	20.6	21.7	12.3	22.4
	20.0	20.1	20.9	29.0	22.2	24.2	32.4 23.0
Zn	19.0 58.0	20.0 50 1	20.3	20.0 62.7	68 5	24.2 81 5	23.0 60 7
Ga	12 9	13.0	13.4	13.5	14.3	14.8	14.4
Rh	85.2	90.5	87.6	93.9	93.6	103	98
Sr	275	420	267	343	335	237	297
Y	24.2	27.2	23.1	23.5	23.0	26.0	27.0
Zr	220	286	230	217	215	231	238
Nb	11.3	11.9	12.0	12.0	12.5	13.2	12.9
Мо	0.99	1.03	0.97	1.90	0.83	1.15	1.17
Sn	2.40	2.35	2.43	2.45	2.67	2.62	2.58
Cs	5.75	6.11	6.07	6.68	7.33	7.90	7.46
Ва	473	503	498	506	481	507	516
La	30.2	32.5	32.1	31.9	32.5	35.3	35.2
Ce	61.1	66.0	64.4	64.8	65.3	71.1	71.5
Pr	7.06	7.60	7.40	7.37	7.51	8.15	8.10
Nd	26.7	28.2	28.1	27.5	28.4	29.7	29.9
Sm	5.38	5.65	5.47	5.56	5.60	5.93	5.87
Eu	1.07	1.06	1.08	1.07	1.10	1.12	1.16
Gd	4.93	5.31	4.92	5.04	5.04	5.37	5.44
Tb	0.71	0.79	0.70	0.74	0.71	0.81	0.82
Dy	4.38	4.57	4.33	4.08	4.40	4.55	4.67
Ho	0.88	0.96	0.85	0.84	0.87	0.94	0.96
Er	2.66	2.82	2.52	2.38	2.54	2.67	2.75
Tm	0.38	0.46	0.37	0.36	0.37	0.41	0.42

Yb	2.52	2.97	2.37	2.31	2.46	2.62	2.64
Lu	0.38	0.47	0.36	0.35	0.37	0.40	0.41
Hf	5.96	7.15	6.25	5.54	5.93	5.95	6.09
Та	0.85	0.92	0.94	0.93	0.96	1.12	0.99
W	1.85	1.82	1.85	2.09	2.07	2.22	2.18
Pb	17.1	18.1	17.7	18.4	19.1	20.4	20.4
Th	10.4	12.3	10.9	11.4	11.4	12.5	12.2
U	3.45	4.55	3.17	4.56	3.05	3.52	3.34

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Supplementary Table 2 Strontium isotope composition of loess and paleosol samples from the Chinese Loess Plateau

			Rb	Sr	Rb/Sr	⁸⁷ Rb/ ⁸⁶ Sr	⁸⁷ Sr/ ⁸⁶ Sr	1 sigma
			ppm	ppm				
	_	16QA-01a	81	293	0.2770	0.8018	0.713080	0.000006
	an	16QA-02a	70	325	0.2146	0.6212	0.712474	0.000006
	Öi	16QA-04a	98	321	0.3060	0.8859	0.713980	0.000003
_		16QA-05a	100	299	0.3328	0.9635	0.714082	0.000006
-		16LC-03a	94	200	0.4728	1.3689	0.715690	0.000005
		16LC-07	108	154	0.7021	2.0335	0.718482	0.000005
		16LC-10a	103	175	0.5855	1.6956	0.717635	0.000006
SSC		16HK-01a	90	313	0.2867	0.8301	0.713878	0.000005
Lec	JL	16HK-05B	92	253	0.3639	1.0535	0.714935	0.000006
	ST 6	16YA-01a	88	229	0.3830	1.1091	0.716307	0.000005
	late	16LT-01a	92	287	0.3211	0.9297	0.714332	0.000005
	ð	16LT-03a	111	181	0.6151	1.7813	0.717137	0.000006
		16LT-04A	102	167	0.6102	1.7672	0.716921	0.000004
		JZT16-06a	88	267	0.3275	0.9483	0.714537	0.000005
		JZT16-07a	94	343	0.2737	0.7924	0.713780	0.000004
		JZT16-08a	94	335	0.2791	0.8080	0.714134	0.000005
		16LC-02a	114	174	0.6528	1.8904	0.716692	0.000005
		16LC-04a	116	183	0.6354	1.8400	0.717166	0.000005
0	8	16LC-06a	117	145	0.8072	2.3380	0.719053	0.000005
e O	2	16LC-08a	114	156	0.7335	2.1244	0.718371	0.000005
Ба	-	16LC-11a	105	139	0.7525	2.1795	0.718775	0.000004
		16LTN-04a	118	172	0.6861	1.9871	0.717532	0.000005
		16LTN-05a	117	138	0.8469	2.4531	0.718856	0.000005

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			Chine	ese Loess Pla	teau (CLF	') - Xining	loess								;LP - X
	XN-2	XN-4	XN-10	XN-6	XN-2	XN-3	XN-4	XN-5	XN-6	XN-10	XF-10 ChauwalEtA	XF-6	XF-2 ChouwolEt	XF-4	XF-
	12014FPSI	2014FPSI	2014EPSI	2014EPSI	JannetA	JannetA	JannetA	2001CG	2001CG	JannetA	12014EPSI	AI2014EPS	AI2014EPS	12014EPSI	. Janne 1200'
SiO	2	20112102	201121 02	20112102	64.45	63.80	66.03	63.27	65.35	65.07			/	12011121 02	68.2
TiO	2				0.65	0.66	0.63	0.71	0.65	0.64					0.70
AI20	03				12.49	12.67	12.01	13.24	12.35	12.31					13.4
Fe20	03				4.61	4.76	4.30	5.12	4.51	4.39					5.00
Mn	C				0.09	0.09	0.09	0.10	0.09	0.09					0.1
Mg	C				2.69	2.74	2.57	2.97	2.62	2.73					2.1
Ca	C				10.46	10.70	9.95	9.93	10.04	10.30					5.6
Na2	0				1.85	1.84	1.85	1.79	1.76	1.84					1.8
K20	C				2.58	2.59	2.42	2.69	2.49	2.47					2.5
P2C	05				0.15	0.15	0.14	0.18	0.14	0.16					0.1
LO Tota	l al				100.02	100.00	99.99	100.00	100.00	100.00					100
	07.5	20.0	22.4	00 7							40.0	00.0	00.0	05.5	
LI	37.5	30.3	38.1	39.7							40.0	39.9	38.0	35.5	
Бе															
50	114	10.9	10.8	11.6							12.4	12.8	12.2	11.5	
V	74.2	69.5	70.0	75.3							84.1	87.1	83.0	76.7	
Cr	64.9	66.8	60.5	70.6							76.9	79.9	73.2	70.7	
Cc	10.9	10.1	10.1	11.1							12.3	13.0	12.1	11.0	
Ni	28.8	26.2	27.2	29.3							34.4	35.9	33.8	32.1	
Cu	22.6	19.7	19.5	21.7							25.3	26.1	23.8	34.1	
Zn	65.9	59.2	60.2	66.7							71.2	77.2	68.8	72.2	
Ga	1														
Ge	•														
As															
Rb	102	95.7	97.2	103	90	89	79	94	67	87	107	112	108	97.4	7
Sr	296	322	303	465	266	250	282	244	386	276	205	232	178	242	1
Y	28.1	28.8	27.2	28.4	22.10	14.50	20.90	23.10	21.40	19.90	31.0	31.1	29.5	29.4	27
Zr	208	269	237	282	122	146	144	136	142	117	238	245	155	250	2
Nb	12.3	12.4	12.3	13.2	11.30	11.30	11.10	11.90	11.10	10.40	13.8	14.5	13.1	13.1	11
Mo)														
Co															
In															
Sn															
Sb		0.05				0.40	5 00					7.00	7.04	0.00	
Cs	6.89	6.35	6.60	6.98	6.20	6.40	5.20	6.80	3.90	6.00	7.90	7.98	7.61	6.66	6.
Ва	481	488	474	510	450	385	460	462	469	470	487	515	495	4/3	4
La	31.0 61.6	30.4 61.2	31.0 62.0	31.1 64.1	20.7	Z1.Z	23.0	30.Z	40.2	24.0 51.6	32.1 66 9	33.9 70.7	34.5	32.1 65.6	
Dr	7 /1	7.07	7 39	7 35	59.5 6.71	02.4 4 73	40.0 5.38	6 00	40.3 5.37	5.83	7 74	8.06	8.26	7.62	7
No	27.4	26.5	27.1	27.5	25.80	4.73	20.80	26.80	20.70	22 30	20.1	29.7	30.1	28.1	30
Sm	1 <u>5</u> 30	5 45	5.34	5 44	4 85	3 25	4 04	5 15	4 04	4 30	5.66	5 90	5.97	5.62	5
Fu	1 00	1 00	0.98	1.05	1 00	0.64	0.87	1 00	0.88	0.90	1 13	1 16	1 12	1.07	1
Go	4.73	4.62	4.68	4.63	4.71	3.17	4.13	4.90	4.16	4.24	5.06	5.19	5.29	4.99	5
Tb	0.761	0.74	0.732	0.753	0.65	0.42	0.58	0.68	0.58	0.57	0.804	0.818	0.821	0.767	0
Dy	4.31	4.28	4.12	4.36	3.85	2.52	3.59	3.98	3.62	3.50	4.62	4.84	4.67	4.49	4.
Hc	0.872	0.887	0.855	0.885	0.76	0.50	0.73	0.79	0.74	0.69	0.95	0.994	0.929	0.921	0
Er	2.58	2.63	2.53	2.59	2.11	1.39	2.04	2.38	2.07	2.01	2.82	2.86	2.66	2.67	2
Tm	ı				0.31	0.20	0.30	0.33	0.31	0.29					0
Yb	2.39	2.44	2.32	2.48	1.98	1.34	1.95	2.06	1.96	1.88	2.57	2.71	2.54	2.51	2
Lu	0.371	0.387	0.365	0.39	0.31	0.21	0.31	0.34	0.32	0.30	0.41	0.416	0.385	0.394	0
Hf	4.88	6.23	5.68	6.62	3.10	3.70	3.70	3.50	3.60	3.00	5.75	5.84	4.06	5.89	6
Та	1.14	0.87	0.88	0.978	0.87	0.95	0.96	0.92	0.83	0.79	0.98	1.02	1.01	0.921	(
W															
TI	0.669	0.657	0.676	0.729							0.746	0.785	0.735	0.676	
Pb	18.7	17.9	18.1	19.4	16.50	16.60	16.30	18.90	17.00	15.50	20.1	20.7	20.6	18.7	1
Bi															
Th	11.6	10.8	11.4	11.3	10.30	7.30	8.40	10.00	8.50	8.50	12.3	12.5	12.6	12.1	1
U	2.94	3.10	2.80	3.66	2.71	3.15	2.91	3.88	3.22	2.52	2.60	2.63	2.35	2.48	2

ļ	g loess								-	CLP - J	ixian loess				
	XF-3	XF-4	XF-5	XF-6	XF-10	JX-2	JX-3	JX-4	JX-5	JX-6	JX-10	JX-2	JX-4	JX-6	JX-10
	JahnEtAl2														
SiO2	64.63	65.48	64.89	64.62	64.29	66.69	65.43	64.59	65.89	66.57	63.83	AIZU 14EF	AIZU 14EF	AIZU 14EF	AIZUI4EF
TiO2	0.68	0.69	0.70	0.72	0.73	0.66	0.73	0.70	0.74	0.70	0.61				
AI2O3	12.96	12.76	13.28	13.50	13.61	12.49	13.60	13.06	13.64	12.61	11.56				
Fe2O3	4.85	4.75	5.03	5.15	5.25	4.48	5.18	4.96	5.20	4.63	4.09				
MnO	0.10	0.09	0.10	0.10	0.10	0.09	0.10	0.09	0.10	0.09	0.08				
MgO	2.42	2.40	2.47	2.63	2.48	2.18	2.49	2.39	2.30	2.40	1.93				
CaO	10.10	9.51	9.13	8.85	9.07	9.13	7.95	9.83	7.66	8.66	13.98				
Na2O	1.65	1.73	1.70	1.61	1.65	1.76	1.67	1.64	1.64	1.77	1.58				
K2O	2.46	2.44	2.55	2.63	2.63	2.37	2.67	2.56	2.67	2.42	2.20				
P2O5	0.16	0.15	0.16	0.18	0.18	0.14	0.18	0.17	0.18	0.15	0.14				
LOI															
Total	100.01	100.00	100.01	99.99	99.99	99.99	100.00	99.99	100.02	100.00	100.00				
Li												33.8	37.1	37.4	30.5
Ве															
В															
Sc												11.4	12.0	11.8	10.8
V												74.3	81.5	77.2	70.0
Cr												69.2	74.0	74.4	64.4
Co												10.4	11.8	11.1	9.82
Ni												27.9	31.8	32.0	26.5
Cu												19.4	22.8	22.2	18.2
Zn												61.1	66.7	64.3	54.8
Ga															
Ge															
AS	00	01	04	77	01	77	01	01	00	00	70	02.7	102	07.0	0E /
Sr	210	232	94 215	203	188	212	167	174	90 156	00 212	213	235	102	97.9	246
v	25.90	27.00	27 30	25 30	24 50	23.60	22.00	21.40	25.90	25 50	19 70	200	29.6	28.0	240
7r	225	239	280	161	139	20.00	103	105	185	213	137	23.0	23.0	262	262
 Nb	11 00	11 70	11 40	11.80	12.30	10 10	11.00	11.30	10.90	11.00	9.90	12.8	13.9	13.6	12.0
Mo					12.00	10.10			10.00		0.00	12.0	10.0	10.0	12.0
Cd															
In															
Sn															
Sb															
Cs	6.30	6.30	6.70	6.60	6.60	5.40	6.30	6.70	6.90	6.20	4.60	6.30	7.43	6.79	5.65
Ва	477	483	481	462	484	432	478	448	464	448	389	442	464	459	414
La	30.7	31.8	31.8	29.4	30.9	29.1	32.6	31.0	31.5	30.8	26.5	30.0	32.2	32.3	27.5
Ce	63.7	66.4	66.4	63.6	64.7	60.3	67.3	63.8	65.9	64.0	54.6	60.6	65.0	65.3	56.9
Pr	7.27	7.52	7.43	7.05	7.21	6.80	7.59	7.20	7.41	7.18	6.19	7.16	7.57	7.58	6.57
Nd	27.80	28.80	28.60	27.30	27.70	26.30	29.10	27.70	28.40	27.50	23.70	26.7	28.2	28.3	24.3
Sm	5.29	5.39	5.34	5.21	5.16	4.90	5.36	5.17	5.33	5.19	4.46	5.25	5.61	5.56	4.67
Eu	1.08	1.09	1.10	1.09	1.10	1.01	1.08	1.05	1.11	1.05	0.91	1.05	1.09	1.05	0.92
Gd	5.00	5.14	5.06	5.08	5.10	4.62	4.95	4.68	5.06	4.92	4.13	4.67	4.97	4.87	4.22
Tb	0.73	0.74	0.73	0.70	0.70	0.67	0.69	0.66	0.73	0.72	0.58	0.745	0.79	0.753	0.701
Dy	4.42	4.61	4.49	4.27	4.21	4.05	3.96	3.84	4.44	4.34	3.52	4.33	4.41	4.35	3.95
Ho	0.89	0.92	0.91	0.86	0.85	0.83	0.78	0.76	0.90	0.88	0.69	0.908	0.908	0.900	0.838
Er	2.54	2.55	2.55	2.38	2.33	2.30	2.15	2.11	2.53	2.49	1.94	2.65	2.63	2.63	2.42
Tm	0.37	0.38	0.39	0.35	0.35	0.34	0.31	0.31	0.37	0.36	0.28				
Yb	2.40	2.53	2.59	2.33	2.27	2.26	2.07	2.03	2.44	2.40	1.84	2.44	2.48	2.50	2.35
Lu	0.40	0.40	0.39	0.36	0.35	0.35	0.32	0.32	0.39	0.38	0.30	0.389	0.387	0.379	0.365
Hf	5.60	5.90	6.80	4.20	3.70	5.20	2.80	2.80	4.80	5.50	3.60	5.65	5.56	6.28	6.14
Та	0.85	0.94	0.84	0.88	0.92	0.83	0.89	0.93	0.90	6.34	0.79	0.904	0.97	0.943	0.837
W															
TI												0.657	0.718	0.694	0.655
Pb	16.40	17.10	17.60	17.40	18.40	16.10	18.30	17.60	17.80	16.80	14.30	17.1	18.7	18.1	16.1
Bi	10 70	14.00	11.00	10 50	10 50	0.70	11.10	10.50	10.00	10.00	0.40	10.0	40.0		0.00
in 	10.70	11.20	11.20	10.50	10.50	9.70	11.10	10.50	10.90	10.30	9.10	10.9	12.0	11.4	9.69
<u> </u>	2.29	2.52	2.53	2.35	2.40	2.31	2.20	2.14	2.28	2.49	2.20	2.30	2.43	2.60	2.37

1																	
2					CI	P - Luoch	uan loess							(CLP - Xinir	ng paleosc)
3		L9	6—4	27—3	40—2	51—1	59—3	84—3	95—3	107—2	115—2	XN-1	XN-7	XN-8	XN-9	XF-1	XF-7
4		ChauvelEt	GalletEtA	GalletEtA	GalletEtA	GalletEtA	GalletEt	GalletEt	GalletEtA	GalletEtA	GalletEt	JahnEtAl	JahnEtAl	JahnEtAl	JahnEtAl	JahnEtAl	JahnEtAl
5	SiO2	AI2014EPS	58 51	58 79	58 20	59.85	57 52	57.80	60.36	11996CG	AI1996C	62 36	2001CG 64.25	2001CG	2001CG	2001CG	2001CG
6	TiO2		0.63	0.64	0.64	0.68	0.64	0.64	0.72	0.66	0.79	0.66	0.71	0.68	0.74	0.72	0.73
7	AI2O3		11.88	12.07	11.92	12.53	12.05	11.97	13.22	12.32	14.29	12.82	13.08	12.72	13.87	13.07	13.54
/	Fe2O3		4.49	4.57	4.55	4.82	4.59	4.57	5.15	4.70	5.65	4.80	4.98	4.87	5.29	4.87	5.19
8	MnO		0.08	0.09	0.08	0.09	0.08	0.09	0.10	0.08	0.11	0.09	0.10	0.09	0.11	0.10	0.10
9	MgO		2.19	2.12	2.07	2.12	2.07	2.00	2.16	2.19	2.21	2.60	2.77	3.02	2.59	2.27	2.51
10	CaO		8.53	8.23	8.31	6.67	8.69	8.49	6.01	7.27	2.89	12.22	9.49	12.62	6.14	6.96	8.89
11	Na2O		1.59	1.53	1.45	1.42	1.37	1.37	1.30	1.37	1.38	1.67	1.85	1.66	1.91	1.73	1.66
12	K2O		2.32	2.29	2.30	2.41	2.26	2.34	2.57	2.39	2.85	2.57	2.61	2.55	2.74	2.70	2.63
12	P2O5		0.15	0.14	0.16	0.14	0.15	0.17	0.16	0.16	0.12	0.20	0.16	0.16	0.16	0.16	0.18
13	LOI		9.26	9.21	9.20	8.17	9.78	9.57	8.19	8.95	6.25	12.57	9.35	11.63	7.65	7.79	8.78
14	Total		99.63	99.68	98.88	98.90	99.20	99.01	99.94	99.80	99.71	100	100	100	100	100	100
15																	
16	Li	37.9															
17	Be																
17	В																
18	Sc	10.9															
19	V	70.8															
20	Cr	63.7															
21	Со	10.2															
21	NI	28.4															
22	Cu	20.8															
23	Zn O-	12.1															
24	Ga																
25	Ge																
26	AS	08	111	00	06	01	09	04	101			00	47	66	100	00	100
20	Sr	170	236	208	108	177	188	170	205			259	215	249	258	188	100
27	Y	27.7	200	200	150	177	100	110	205			21.80	20.80	16.80	230	26 50	28 30
28	Zr	216	205	182	192	206	163	190	186			119	139	167	129	231	219
29	Nb	12.1	10.8	6.8	11.6	12.3	10.6	12.3	11.8			11.20	11.80	11.20	12.20	11.60	12.00
30	Мо																
31	Cd																
27	In																
52	Sn																
33	Sb																
34	Cs	7.73	8.20	7.20	7.20	7.30	7.30	7.10	7.20			7.10	2.50	4.80	8.80	6.30	7.20
35	Ва	459	545.00	452.00	441.00	436.00	446.00	429.00	442.00			447	441	425	504	479	490
36	La	31.7	38.10	32.10	31.50	32.20	30.30	30.90	31.10			29.0	21.4	18.7	33.2	32.5	32.2
37	Ce	67.2	75.50	63.70	62.80	64.00	61.10	62.40	62.50			60.3	47.9	45.2	69.7	68.1	66.6
20	Pr	7.81	9.34	8.20	8.07	7.96	7.75	7.91	7.87			6.76	5.24	4.26	7.79	7.72	7.52
38	Nd	29.6	32.50	28.00	27.70	27.70	26.60	27.30	27.20			25.90	20.30	16.20	29.70	29.50	28.90
39	Sm	5.74	6.61	5.60	5.57	5.76	5.33	5.48	5.52			4.85	3.99	3.18	5.54	5.58	5.42
40	Eu	1.13	1.31	1.13	1.10	1.11	1.05	1.04	1.07			0.99	0.85	0.67	1.12	1.13	1.13
41	Gd	5.06	5.91	5.09	5.08	5.17	4.72	4.86	4.92			4.70	4.03	3.32	5.15	5.26	5.18
17	Ib	0.801	0.92	0.78	0.77	0.80	0.74	0.76	0.77			0.64	0.58	0.47	0.72	0.74	0.77
42	Dy	4.62	5.19	4.61	4.57	4.59	4.22	4.38	4.40			3.82	3.63	2.91	4.33	4.61	4.68
43	Но	0.964	1.08	0.92	0.94	0.95	0.85	0.88	0.89			0.76	0.73	0.58	0.85	0.93	0.95
44	Er	2.78	2.97	2.61	2.60	2.72	2.37	2.47	2.52			2.08	2.08	1.66	2.42	2.65	2.69
45	1 M Vh	2 6 2	0.49	0.44 2.74	0.43	0.44 2.70	0.39	0.41	0.41			0.30	0.31	0.24	0.35	0.39	0.40
46	U I	2.03	3.U3 0.4E	2.74	2.12	2.70	2.41	2.55	2.00			1.90	1.97	0.25	2.22	2.53	2.04
47	LU	5.61	0.40 5 34	0.42 4 70	0.40 5 1 1	0.41	0.30 1 30	5.00	0.40 4 QF			U.J I 3 10	3 70	0.20 1 20	3 40	6.00	0.4 I 5 50
4/	Та	0.071	0.04	9.19	0.11	1.06	4.39 0.90	0.09	4.90			0.10	0.03	4.20 0.84	0.40	0.00	0.00
48	1 d \\\/	0.971	0.07	0.47	0.09	1.00	0.00	0.90	0.94			0.09	0.93	0.04	0.91	0.09	0.90
49	VV TI	0 755															
50	Ph	10.5	23.00	20.00	20.00	23.00	20.00	20.00	21.00			17 20	16.80	16.00	19 10	17 20	18 20
51	Ri	18.5	20.00	20.00	20.00	20.00	20.00	20.00	21.00			17.20	10.00	10.00	13.10	11.20	10.20
51	Th	11.6	14.30	11.90	11.90	13.00	11.20	11.50	11.90			10.30	8,10	7.60	11.90	10.90	11.50
52	U	2.52	2.93	2.94	2.84	2.69	2.58	2.66	2.65			2.38	3.09	3.24	3.26	2.46	2.46
F 2	-	-		-	-												

International Geology Review

	CLP - Jixian paleosol					bl			CLP - L	uochuan p	aleosol				
	XF-8	XF-9	JX-1	JX-7	JX-8	JX-9	18—1	32—3	47—3	58—3	71—2	91—2	97—2	LO94	P2E1
	JahnEtAl	JahnEtA	JahnEtA	JahnEtA	JahnEtAl	JahnEtAl	GalletEtAl	GalletEtAl	GalletEtAl	GalletEtAl	GalletEtAl	GalletEtAl	GalletEtAl	ChauvelEtA	ChauvelEt/
0:00	2001CG	12001C	12001C	12001C	2001CG	2001CG	1996CG	1996CG	1996CG	1996CG	1996CG	1996CG	1996CG	I2014EPSL	I2014EPSI
5102	66.19	64.13	/1.25	68.31	70.12	67.96									
1102	0.71	0.72	0.79	0.81	0.85	0.80									
AI2U3	13.26	13.67	14.55	14.30	14.74	14.23									
Fe2O:	4.99	5.28	5.40	5.49	5.64	5.44									
MaQ	0.10	0.10	0.11	0.10	0.10	0.10									
CaO	2.35	2.40 0.18	1.79	4.22	2.10	2.25									
Na2O	1.66	1.64	1.52	1.61	1 40	1.60									
K20	2.57	2.63	2 73	2 72	2.68	2.69									
P205	0.15	0.18	0.15	0.14	0.11	0.14									
1.01	8 4 1	9.10	4 17	6 24	5 11	6.63									
Total	100	100	100	100	100	100									
Li														61.3	59.9
Be															
в															
Sc														8.73	8.63
V														85.2	83.5
Cr														69.4	67.4
Co														8.85	8.96
Ni														25.3	26.7
Cu														17.1	17.6
Zn														108	69.4
Ga															
Ge															
As															
Rb	94	83	95	97	65	102	86.00	88.00	53.00	93.00	41.00	96.00	95.00	63.7	65.6
Sr	194	193	126	145	123	151	112.00	135.00	117.00	123.00	64.00	153.00	178.00	133	135
Y	26.30	25.20	27.70	24.40	24.40	26.60	31.50	32.20	30.50	32.00	19.60	53.50	29.80	21.8	21.7
Zr	201	119	222	118	190	126	207.00	210.00	214.00	230.00	184.00	171.00	159.00	213	263
Nb	11.50	12.10	11.40	11.20	13.20	13.40	15.10	9.90	12.70	10.90	6.70	12.70	11.90	11.6	12.3
Мо															
Cd															
In															
Sn															
Sb		- 10					0.40	0.40	7.00	7.00	- 10		7.40	o o -	
Cs D-	6.40	7.40	7.50	7.70	7.70	7.80	9.10	9.10	7.60	7.90	7.10	7.70	7.40	3.87	3.73
ва	4/3	496	502	501	499	502	198.00	494.00	396.00	494.00	412.00	454.00	434.00	510	557
La	31.Z	33.3	34. I 74. 0	34.9	27.4	34.2	33.30	54.20	31.40	35.10	15.60	49.10	34.00	27.5 61.5	20.4
Dr	7 22	09.0	71.0	72.0	6 71	/ I. I 9.06	47.20	52.3U	47.00	01.10	30.10	11 01	02.30	C.10	6.62
Nd	7.33	20.80	30.60	30.00	26.00	30.00	0.09	0.92 31.40	28 70	30.50	4.32	11.91	20.50	25.5	24.6
Sm	5 26	5 53	5 74	5.80	5.03	5 78	6.09	6 38	5.80	6.40	3 32	9.84	5 79	4.86	4 70
Eu	1.08	1 16	1 20	1 16	1.02	1.21	1.21	1 25	1 12	1.22	0.67	2.04	1 17	4.00	4.70
Gd	5.05	5 14	5.34	5 19	4 71	5.28	5.48	5.68	5.22	5.63	3.21	9.86	5.24	4 20	4.06
Th	0.00	0.74	0.80	0.73	0.70	0.20	0.85	0.89	0.81	0.87	0.51	1.52	0.80	0.647	0.625
Dv	4 40	4.39	4 81	4 27	4 25	4 67	4 93	5 17	4 77	4 87	3.20	8.71	4 68	3.88	3 76
Ho	0.89	0.88	0.97	0.85	0.86	0.93	0.99	1.04	0.96	1.02	0.66	1.78	0.95	0.818	0.787
Er	2.49	2.43	2.74	2.35	2.43	2.62	2.79	2.93	2.69	2.96	1.91	4.87	2.62	2.43	2.34
Tm	0.38	0.36	0.40	0.34	0.36	0.38	0.47	0.49	0.45	0.46	0.32	0.73	0.44		
Yb	2.49	2.39	2.62	2.27	2.38	2.45	2.97	3.06	2.84	2.93	2.11	4.38	2.73	2.42	2.28
Lu	0.39	0.36	0.41	0.35	0.38	0.38	0.45	0.48	0.45	0.44	0.32	0.62	0.40	0.378	0.353
Hf	5.20	3.30	5.60	3.20	5.00	3.40	5.60	5.72	5.84	6.15	4.98	4.60	4.17	5.92	6.98
Та	0.87	0.95	0.90	0.94	1.09	1.03	1.19	0.85	1.04	1.00	0.35	0.99	0.87	0.881	0.901
W															
TI														0.569	0.585
Pb	17.60	18.80	20.00	19.60	20.50	19.40	22.00	22.00	21.00	23.00	15.00	21.00	21.00	15.4	15.7
Bi															
Th	10.90	11.70	11.70	11.80	9.60	12.20	12.80	12.60	12.40	14.00	55.00	13.10	11.40	7.85	7.64
	0.44	2 21	2 30	2 33	2 55	2 40	2.06	2.27	2.15	2.66	1.10	2.58	2.47	2 19	2 11

	Sp	oitsbergen ((Svalbard)				Engl	and		Belgium				
	P2E1	P2E2	P2E3	LO94	GAI	2408M3	SCIL	SCIL	K14	R11	R11dup	PR	PR-RT	HOT
	GalletEtAl	GalletEtAl	GalletEtAl	GalletEtAl	GalletEtAl	GalletEtAl	ChauvelEtA	GalletEtAl	ChauvelEtA	ChauvelEtA	ChauvelEtA	GalletEtAl19	ChauvelEtA	GalletEtAl1
SiO2	76.66	76.85	76.65	78.73	78.66	76.98	1201411 01	83.94	120141101	120141101	120141101	77.72	120141101	75.59
TiO2	0.70	0.74	0.75	0.61	0.60	0.72		0.57				0.44		0.70
AI2O3	12.50	12.32	12.41	10.85	10.94	12.17		8.50				6.23		8.34
Fe2O3	4.80	4.75	4.91	3.52	3.51	4.83		2.80				1.96		2.99
MnO	0.04	0.03	0.03	0.04	0.04	0.03		0.05				0.04		0.04
MgO	0.77	0.75	0.74	1.04	1.03	0.72		0.59				1.11		1.42
CaO	0.45	0.41	0.43	0.56	0.55	0.43		0.46				9.71		7.83
Na2O	1.43	1.54	1.41	2.20	2.22	1.56		0.96				0.97		1.12
K2O	2.49	2.44	2.49	2.37	2.36	2.40		2.05				1.70		1.86
P205	0.15	0.17	0.17	0.10	0.09	0.16		0.07				0.11		0.11
LOI														
Total														
Li							41.1		30.9	44.6	44.5		17.9	
ве														
В							6.46		10.0	10.2	10 F		6.01	
SC							0.40		10.9	10.3	10.5		0.01	
v							39.Z		55.9	20.3	00.0 110		20.4	
Co							6.01		95.4	7.00	0.05		50.9	
Ni	25.00	20.00	27.00	27.00	10.00	27.00	18.0	20.00	0.79 21.0	7.90	0.00	15.00	4.20	10.00
Cu	25.00	29.00	27.00	27.00	10.00	27.00	7.62	20.00	10.2	10.2	10.3	15.00	6.38	19.00
Zn							42.8		58.0	54.0	43.3		25.3	
Ga							-12.0		00.0	04.0	40.0		20.0	
Ge														
As														
Rb	81.00			82.00		77.00	68.4		84.6	71.8	73.8	52.00	47.4	62.00
Sr	159.00			156.00		157.00	64.2	75.00	83.2	66.5	68.2	197.00	190	174.00
Y	23.70			22.60		23.00	16.9		34.1	31.5	32.1	18.10	19.3	29.20
Zr	303.00	272.00	291.00	243.00	261.00	300.00	356	327.00	593	620	622	307.00	343	367.00
Nb	15.00			14.00		13.50	10.2		15.9	17.6	17.5	7.90	7.77	11.70
Мо														
Cd														
In														
Sn														
Sb														
Cs							4.28		3.83	3.51	3.53		2.18	
Ва	567			519		536	292		373	350	352	246	245	297
La	29.70			29.30		28.80	19.0	19.66	34.5	34.9	34.3	19.5	19.7	26.7
Ce	63.40			62.30		60.40	43.2	42.95	76.3	76.4	74.4	39.6	41.3	54.6
Pr	7.70			7.70		7.40	4.70	4.40	9.04	8.61	8.50	5.30	5.10	7.10
Nd	26.90			26.30		25.60	16.9	16.83	34.4	32.2	31.3	18.70	19.3	25.0
Sm	5.20			5.02		4.84	3.20	3.14	6.83	6.16	5.81	3.78	3.9	5.06
Eu	1.10			1.07		1.05	0.592	0.57	1.26	1.08	1.08	0.72	0.667	0.98
Gd	4.44			4.24		4.24	2.75	3.00	5.99	5.34	5.25	3.41	3.33	4.62
Tb	0.69			0.66		0.65	0.445	0.41	0.938	0.842	0.831	0.51	0.518	0.74
Dy	3.87			3.65		3.68	2.74	1.98	5.52	5.14	5.17	2.90	3.05	4.41
Ho	0.81			0.76		0.78	0.593	0.40	1.16	1.10	1.11	0.58	0.626	0.92
Er	2.16			2.09		2.13	1.77	1.09	3.33	3.24	3.35	1.56	1.85	2.59
l m	0.36			0.34		0.35	4.00	1.18	0.00	0.00	0.07	0.25	1.00	0.44
YD	2.31			2.15		2.24	1.86	0.40	3.38	3.33	3.37	1.66	1.82	2.81
LU LIF	0.30			0.34		0.30	0.295	0.18	0.524	16.2	0.537	U.25	U.∠86 g ⊑	0.43
	1.00			0.00		0.00	9.02		14.0	10.3	10.1	0.60	0.0	1 40
18	1.00			0.90		0.90	0.007		1.23	1.41	1.37	0.60	0.047	1.40
VV TI							0 501		0 665	0.617	0.600		0 412	
Ph	18 80			18 10		17 00	14 1		16.6	15.8	16.0	10 70	10.0	13 10
Ri	10.00			10.10		11.50	14.1		10.0	10.0	10.0	10.70	10.0	13.10
Th	9 70			8 70		8 50	7 40		9 99	10 5	10 7	6 40	6 4 4	8 30
	5.10			5.10		0.00	1.40		0.00	10.0		0.40	0.1-1	0.00

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				Fra	nce							
	HOT1-WholeRock	HOT1<160mm	FOUG3	FOUG3a<160mm	HC90/1	SAB1	SAB1a<160mm	NS2	NS3	NS4	NS6	NS6<160mm
	ChauvelEtAl201	ChauvelEtAl20	GalletEtAl	ChauvelEtAl2014	GalletEtAl1	GalletEtAl1	ChauvelEtAl20	GalletEtAl	GalletEtAl1	GalletEtAl1	GalletEtAl	ChauvelEtAl2
SiO2	4EPSL	14EPSL	1998EPS	EPSL	998EPSL	998EPSL	14EPSL	1998EPSL	998EPSL	998EPSL	1998EPSL	014EPSL
3102 TiO2	<u>-</u>		04.74		0.78	0.37		0.79	0.78	0.93	0.97	
AI20	- 3		8.51		8.23	6.45		9.84	11.09	10.59	10.65	
Fe2O	3		2.47		3.20	1.83		4.34	4.42	5.05	4.51	
MnC)		0.02		0.05	0.03		0.06	0.07	0.08	0.06	
MgC)		0.36		1.16	0.99		0.92	0.88	1.21	1.29	
CaO)		0.28		7.76	7.00		21.46	1.14	7.05	3.93	
Na20	D C		0.93		0.78	0.99		1.59	1.89	1.56	1.51	
K20	1		1.76		1.92	1.70		1.55	2.16	1.82	1.89	
P20	5		0.05		0.12	0.09		0.11	0.10	0.14	0.12	
LOI												
Tota	I											
	25.0	22.0		20.6			16.6					20.6
LI Re	20.0	23.9		29.0			10.0					29.0
R												
Sc	8.33	7.71		8.73			4.55					10.6
V	46.8	41.8		43.9			26.0					68.0
Cr	61.3	62.5		76.0			35.4					68.5
Co	6.09	4.87		3.80			3.97					7.66
Ni	16.7	15.3	7.00	10.3	22.00	12.00	11.6	12.00	16.00	24.00	21.00	18.9
Cu	9.04	8.4		4.95			6.33					11.5
Zn	38.5	34.3		38.5			36.7					50.7
Ga												
Ge												
As												
Rb	58.3	56.8	62.00	61.0	64.00	54.00	50.8				68.00	64.5
Sr	167	167	59.00	62.8	150.00	152.00	170				131.00	131
Y	27.4	24.4	20.60	33.1	28.80	15.10	14.0				28.00	31.1
Zr	408	387	468.00	562	487.00	223.00	193	149.00	161.00	305.00	342.00	411
ND	10.9	10.2	14.70	15.4	8.60	6.10	5.78				12.40	11.8
Cd												
In												
Sn												
Sb												
Cs	3.25	2.90		3.03			1.93					3.48
Ba	301	281	301	299	282	277	277				329	324
La	25.2	23.4	28.4	28.2	29.5	14.8	13.8				30.4	28.3
Ce	53.6	47.6	60.7	67.7	59.7	29.9	28.4				60.5	60.5
Pr	6.43	5.98	7.30	7.30	7.80	4.00	3.58				8.00	7.20
Nd	24.5	22.7	25.6	27.4	27.5	14.3	13.9				28.4	27.6
Sm	4.94	4.48	4.97	5.38	5.59	2.99	2.70				5.67	5.55
Eu	0.938	0.852	0.83	0.854	1.03	0.61	0.562				1.13	1.07
Gd	4.44	4.07	4.14	4.80	4.99	2.68	2.35				5.04	5.01
Tb	0.706	0.64	0.64	0.784	0.78	0.41	0.362				0.79	0.805
Dy	4.31	3.91	3.50	4.90	4.53	2.43	2.21				4.53	4.98
Ho	0.93	808.0	0.71	1.06	0.93	0.49	0.455				0.89	1.05
Er	2.70	2.39	1.97	3.21	2.53	1.29	1.31				2.49	3.07
i M Vh	2.69	2 33	0.33 2 14	3 23	0.43	0.21 1.40	1 20				0.40 2.50	3.01
10	0 421	0.363	0.33	0.511	0.42	0.21	0 199				0.30	0.461
Hf	10.4	9.51	0.00	13.5	0.42	0.21	4,60				0.55	10.3
Ta	0.892	0.815	1,10	1.20	0.70	0.40	0.473				1,00	0,958
W												
ΤΙ	0.512	0.478		0.572			0.416					0.568
Pb	12.3	11.0	15.50	14.2	13.80	10.70	9.68				14.70	12.5
Bi												
Th	7.73	7.45	9.70	9.26	9.20	4.70	4.34				9.00	8.63
	0.17	2.00	2.54	2 76	2 34	1 1 9	1 16				2 01	2 25

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						Argentir	na, South A	merica					Sahara	, Africa
	4—6 CalletEtA	12—14 CalletEtA	24—26 CalletEtA	32—34 GalletEtA	40—42 GalletEtA	52—54 GalletEtA	LUJA CalletEtA	12-14 ChauwelEt	24-26 ChauvelEt	32RT ChauvelEt	40RT	LUJA	Sample Meylan	Sample 2900m
	I1998EP	I1998EP	I1998EP	I1998EP	I1998EP	I1998EP	I1998EP	AI2014EPS	AI2014EPS	AI2014EPS	AI2014EPS	AI2014EPS	14EPSL	14EPSL
SiO2	65.02	70.66	62.39	65.63	66.94	69.50	70.22							
TiO2	0.68	0.74	0.72	0.89	0.91	1.09	0.67							
AI2O3	15.14	14.41	15.01	16.91	16.20	13.59	15.40							
Fe2O3	4.81	4.34	4.67	5.58	5.64	5.33	3.58							
MaQ	1.54	1 10	1.35	1.67	2 15	1.56	1.36							
CaO	8.41	3.41	10.94	4.19	3.55	4.84	2.60							
Na2O	1.99	2.67	2.44	2.60	1.93	1.50	3.36							
K2O	2.18	2.37	2.12	2.28	2.44	2.14	2.65							
P2O5	0.16	0.18	0.23	0.15	0.14	0.14	0.10							
LOI														
Total														
								00.0	00.4	04.0	00.7	20.0	40.7	50.5
LI								26.3	20.1	31.8	30.7	30.6	48.7	56.5
B														
Sc								10.9	11.9	13.6	14.0	10.0	13.0	12.3
V								82.2	85.9	107	94.8	73.6	102	106
Cr								34.1	28.8	35.6	36.1	30.1	87.3	95.6
Co								9.79	10.3	10.0	10.2	5.85	14.2	13.4
Ni	9.00	39.00	9.00	13.00	14.00	19.00	7.00	71.8	11.4	12.7	15.2	8.77	36.0	40.3
Cu								28.9	23.8	27.0	35.3	16.4	40.1	38.4
Zn								70.1	66.6	69.3	79.1	54.7	138	116
Ga														
Ge														
Rb	76.00	81.00	68.00	76.00	82.00	82.00	90.00	79.5	77.2	64.6	68.8	85.5	87 7	67.4
Sr	261.00	292.00	326.00	371.00	285.00	246.00	330.00	314	375	358	301	330	298	95.5
Y	21.10	22.50	21.80	25.40	23.60	28.10	20.50	25.3	29.3	29.8	26.6	25.6	37.2	33.3
Zr	186.00	252.00	201.00	232.00	225.00	290.00	252.00	257	213	247	253	269	270	202
Nb	8.10	9.70	8.10	9.40	10.70	14.90	9.50	10.1	8.56	10.1	11.1	10.3	22.9	23.3
Мо														
Cd														
In														
Sn														
50								4.62	4.32	4 70	5 76	4 59	4 33	4 95
Ba	504	543	565	600	423	890	524	+.02 555	572	643	480	579	552	363
La	21.3	23.2	21.4	22.5	25.1	30.1	23.5	23.3	22.7	23.9	24.7	21.8	41.8	37.7
Ce	41.0	47.1	42.3	46.5	52.2	69.5	46.8	47.3	45.1	48.9	52.6	45.4	84.7	79.1
Pr	5.60	6.20	5.80	6.40	6.90	8.20	6.40	5.87	5.9	6.31	6.57	5.57	10.1	9.35
Nd	20.4	21.9	21.4	23.4	24.7	29.0	22.6	22.1	22.9	24.8	25.1	21.2	37.9	34.7
Sm	4.23	4.43	4.44	4.89	5.20	5.80	4.46	4.39	4.74	5.18	5.25	4.30	7.28	6.68
Eu	1.01	1.03	1.06	1.25	1.16	1.30	1.02	1.01	1.10	1.22	1.18	0.997	1.49	1.31
Gd ⊤⊾	3.68	3.96	3.90	4.44	4.50	5.12	3.80	3.96	4.19	4.64	4.53	3.79	6.31	5.61
	3 16	3.51	0.09	0.07 4.05	3.81	4.61	0.00	3 71	4 26	4 35	4 20	3.71	0.900	5.11
Ho	0.66	0.75	0.71	0.82	0.78	0.92	0.68	0.788	0.887	0.905	0.849	0.779	1.16	1.04
Er	1.85	2.07	1.92	2.26	2.20	2.51	1.84	2.34	2.6	2.66	2.47	2.36	3.36	2.97
Tm	0.29	0.34	0.30	0.37	0.35	0.41	0.31							
Yb	1.91	2.13	2.00	2.42	2.17	2.69	2.04	2.30	2.51	2.57	2.30	2.32	3.03	2.80
Lu	0.28	0.34	0.30	0.36	0.25	0.40	0.31	0.35	0.394	0.399	0.358	0.361	0.47	0.427
Hf								6.04	4.90	5.75	6.00	6.34	6.53	4.96
Та	0.70	0.80	0.60	0.70	0.90	1.10	0.80	0.732	0.579	0.685	0.758	0.755	1.48	1.67
W								0 700	0.015	0.740	0 700	0.740	0.000	0.770
TI	14.90	20.90	16.00	16 70	10.00	25.90	15.00	0.733	0.645	0.713	0.798	0.743	0.662	0.772
PD Di	14.80	20.80	16.30	10.70	18.20	25.80	15.80	19.7	15.2	13.6	13.8	13.5	26.0	42.2
Th	7.70	9.30	7.40	7.80	9.80	9.40	9.70	9.16	7.75	8.29	9.22	9.48	12.0	12.2
		0.00			0.00	0.40	0.70	0.10		0.20	J.LL	00	.2.0	

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					Tadji	ikistan					Kar	isas	
	TJK2772	TJK2773	TJK2930	TJK3012	TJK3070	TJK3148	TJK3179	TJK3198	TJK2814	TJK3165	CY-4a-A	CY-4a-B	BP-1
	ChauvelEtA	Hu&Gao2	Hu&Gao2	Hu&Gao									
SiO2	12014LF 3L	12014LF3L	12014LF 3L	12014LF 3L	12014LF3L	12014LF3L	12014LF 3L	12014LF 3L	12014LF 3L	12014LF3L	80.4	80.8	72.7
TiO2											0.64	0.63	0.57
AI2O3											10.5	10.6	15.8
Fe2O3	3										2.58	2.64	3.30
MnO											0.04	0.04	0.051
MgO											0.86	0.88	0.95
CaO											1.12	1.23	1.54
Na2O											1.58	1.68	3.27
K2O											2.53	2.62	2.39
P2O5													
LOI													
Total													
Li	44.3	43.0	35.6	46.7	35.4	37.8	55.2	32.5	36.5	45.8	15.7	17.9	32.1
Be											1.65	1.66	2.85
В											42.5	40.3	30.4
Sc	14.8	14.6	12.4	16.0	13.0	13.0	16.4	11.7	13.2	12.7	6.89	6.78	8.4
V	113	111	88.9	123	99.0	98.9	131	82.6	97	117	60	61.2	67
Cr	96.7	95.1	79.9	103	85.7	82.4	107	72.7	87.5	96.9	42.8	42.9	35.9
Со	15.4	15.1	12.5	17.2	13.8	14.0	17.9	12.6	14.0	16.4	5.77	5.97	11.3
Ni	51.6	51.2	42.8	57.3	43.8	43.8	59.9	38.5	44.2	51.7	14	15	15.3
Cu	33.5	34.4	26.8	37.8	29.0	30.1	41.8	24.9	29.5	34.8	12	12.3	14
Zn	89.6	86.3	72.3	98.2	81.3	84.4	111	72.4	79.7	98.3	40.1	40.9	67.9
Ga											11.3	11.5	14.5
Ge											1.43	1.4	1.44
AS	444	110	02.2	02.4	07.7	101	110	01.0	02.6	100	3.87	3.74	4.11
RD Sr	114	200	93.2	93.1	97.7	101	125	91.0	93.0	120	02.1 204	03.5 207	95.7
v	21.0	200	200	25.7	209	220	26.7	229	270	20 6	204	207	214
T Zr	208	206	29.1	232	178	178	232	20.0	104	20.0	589	29.0 573	23.7
Nh	14.2	13.0	12 7	15.9	12.3	12 7	16.6	12.0	13.0	16.1	14.6	15	10.8
Mo	14.2	10.0	12.1	10.0	12.5	12.7	10.0	12.0	10.0	10.1	0.82	0.94	0.34
Cd											0.02	0.04	0.023
In											0.032	0.032	0.042
Sn											1.4	1.43	2.18
Sb											0.73	0.76	0.34
Cs	6.83	6.65	5.32	7.07	5.70	6.22	8.24	5.37	5.57	7.07	2.95	3.05	4.48
Ва	515	492	438	497	472	487	570	446	451	571	679	692	548
La	32.4	32.7	30.9	37.1	28.5	30.4	37.6	29.7	30.1	31.8	39.7	35.2	35.3
Ce	67.5	65.9	61.4	76.2	58.7	60.5	79.5	57.4	62.1	73.5	78.7	69.3	72.9
Pr	7.64	7.77	7.25	9.15	6.84	7.09	8.80	6.84	7.09	7.41	9.48	8.37	8.58
Nd	29.1	29.0	27.0	34.5	25.5	26.4	33.3	25.5	26.6	27.6	35.3	31.3	32.2
Sm	5.670	5.82	5.27	6.97	4.99	5.27	6.51	5.03	5.20	5.24	6.56	5.92	6.13
Eu	1.13	1.19	1.03	1.41	1.02	1.06	1.31	0.97	1.10	1.08	1.27	1.23	1.33
Gd	5.01	5.21	4.68	6.11	4.42	4.58	5.90	4.40	4.69	4.68	5.66	5.13	5.05
Tb	0.807	0.840	0.744	0.969	0.717	0.740	0.965	0.706	0.770	0.765	0.9	0.8	0.74
Dy	4.82	4.82	4.33	5.61	4.10	4.30	5.59	3.99	4.26	4.36	5.49	4.85	4.2
Ho	0.981	1.00	0.880	1.13	0.849	0.875	1.130	0.796	0.887	0.919	1.19	1.02	0.85
Er	2.85	2.92	2.52	3.22	2.44	2.50	3.34	2.30	2.56	2.62	3.5	2.98	2.39
Tm											0.58	0.5	0.38
Yb	2.69	2.74	2.37	3.00	2.24	2.40	3.09	2.18	2.41	2.46	3.66	3.19	2.4
Lu	0.408	0.421	0.364	0.458	0.349	0.360	0.478	0.333	0.375	0.386	0.58	0.51	0.38
Hf	5.02	4.95	4.62	5.58	4.32	4.29	5.55	4.12	4.66	5.18	14.9	14.4	8.85
Та	1.01	1.02	0.930	1.13	0.883	0.923	1.18	0.873	0.890	1.13	1.01	1.05	0.89
W		c = · ·		c	A 4/-	e ====	· · · ·	c	· · · ·		1.26	1.32	1.41
TI	0.751	0.740	0.631	0.787	0.642	0.667	0.892	0.629	0.656	0.857	0.6	0.61	0.64
Pb	21.4	20.6	18.4	18.1	18.7	19.5	24.2	18.6	18.1	22.8	15.3	16.2	15.7
Bi	10.4	14.0	14.5	10.4	10.0	14.0	14.4	14.0	14.0	10.0	0.16	0.16	0.19
	12.1	11.3	C.11	13.1	10.9	11.4	14.1	11.0	11.0	13.3	1.3.4	110	11

		Bank	s Penn.,	New Zea	land		CN		K	aiserstuhl		Hungary		Kansas		Iowa	
	BP	-2	BP-3	BF	P-4	BP-5	Nanking	CH	1	2+		Н	Kansas A	Kansas B	Kansas C	lowa	
	I ayloEtA	Barth20	Hu&Ga	LayloEt	Barth2	Hu&Ga	1 ayloEtAl1 983GCA	Hu&Ga	Hu&Ga	TayloEtA	Barth 2000	Hu&Ga	1983GCA	1983GCA	1983GCA	TayloEtAl	
SiO2	74.00	0000	72.5	74.0	00000	72.5	72.8	020000	59.9	59.1	2000	020000	80.40	80.80	79.90	79.50	
TiO2	0.55		0.69	0.57		0.54	0.78		0.32	0.29			0.64	0.63	0.71	0.69	
AI2O3	3 14.70		15.2	14.30		15.1	15.4	13.18	7.78	7.98			10.5	10.6	10.7	11.40	
Fe2O	3 3.38		3.95	3.45		3.21	4.79	4.70	3.22	2.95			2.58	2.65	2.55	3.01	
MnO	0.054		0.054	0.055		0.052	0.12		0.07	0.070			0.044	0.045	0.043		
	0.97		0.88	1.07		1.06	1.59		3.45 23.11	4.37 22.90			0.86	0.95	0.88	1.01	
Na2C) 3.05		3.13	3 43		3.55	1.28		0.84	0.87			1.12	1.09	1.23	1 45	
K20	2.31		2.26	2.42		2.47	2.21		1.27	1.34			2.53	2.57	2.62	2.18	
P205	5																
LOI													2.32	2.38	2.29		
Total													2.58	2.65	2.55		
Li	31		26.94	35		38.8	38	39.2	21.1			20.8		4.5		21	
ве	2.3		2.43	2.2		2.44	2.4	2.19	1.19	1.1		1.0	1.4	1.5	1.4	1.4	
Sc	84		10.8	80		8 29	11.9	12.6	7 21	6.0		9.22	57	54	49		
V	63		66.5	65		65.1	99	85.2	42.5	39		61.8	55	57	57		
Cr	32		35.4	30		34.2	69	70.1	58.4	42		59.1	31	33	32		
Co	11		8.8	8		7.98	18	12.6	6.62	6		8.37	5	7	5		
Ni	14		12.3	13		15.3	34	35.5	26.2	23		26.6	13	13	11		
Cu	12		14.2	12		13.2	30	25	14.6	13		18	10	12	10		
Zn	54		51.3	53		47.5	78	64.2	36	35		48	46	45	44		
Ga	14	16.70	15.6	15	16.70	14.3	17	14.7	7.19	6.6	6.80	10.4	10	10	10		
Δs			4.21			4.26		1.52	8.51			7 79					
Rb	86.7		84.4	84.2		96.8	108	10.0	47.9	49.7		68.8	74.2		71.9	77.00	
Sr	310	304	318	275	276	323	104	230	396	302	300	195	187	205	195	162	
Y	23	26.5	30.8	23	26.3	21.2	26.0	30.4	25	19.00	28.80	29.1	22.00	21.00	21.00		
Zr	380	410	512	390	401	375	330	218	255	260	293	324	400	390	570	350	
Nb	9.9	11.20	17.4	17	11.8	10.7	24	13.1	8.97	9.3	9.50	13	16	21	21	20	
Мо			0.48			0.4		0.72	0.43			0.44					
Cd			0.026			0.026		0.13	0.13			0.13					
IN Sn	47		2 20	4 9		0.04	8.1	2.63	0.032	4.0		2.22	2.8	3.6	4.4	4 4	
Sb	4.7		0.29	4.5		0.33	0.1	1.46	0.62	4.0		0.91	2.0	5.0	7.7	7.7	
Cs	2.9	4.50	3.89	4.2	4.70	4.4	7.7	7.1	2.71	1.50	2.60	3.55	3.0	2.5	2.3	3.40	
Ва	590	614	566	595	583	551	480	485	212	190	192	300	635	660	620	810	
La	32.0	33.7	41.6	34.6	34.3	34.4	38.5	33	26.5	25.0	25.9	34.2	34.0	31.0	39.0	33.1	
Ce	74.0	75.2	81.6	74.4	72.0	73.1	84.9	66.9	51.5	51.0	52.4	65.9	77.0	77.0	83.0	74.1	
Pr	8.40	8.48	10.1	9.48	8.37	8.08	9.14	7.72	6.18	5.87	6.44	7.88	7.14	6.17	9.55	8.08	
Nd Sm	31.4	30.8 5.09	39.3	31.6	30.2	30.1 5.54	35.2	29.2	24.2	23.2	23.8	30.5	31.7	24.9	33.5	36.0	
Fu	1.90	J.90 1 41	1 54	5.95 1 14	J.70 1.36	J.J4 1 19	1.28	1 25	4.94 0.86	+.00 0.82	0.91	1.08	0.09	4.59 0.86	1 12	0.70	
Gd	5.09	5.64	6.09	4.38	5.54	4.55	4.98	5.19	4.26	3.39	4.76	5.13	4.00	3.15	4.40	4.52	
Tb	0.79	0.79	1	0.70	0.77	0.67	0.92	0.82	0.72	0.62	0.73	0.87	0.63	0.52	0.79	0.88	
Dy	4.38	4.28	5.37	4.38	4.29	3.8	5.48	4.86	4.06	3.89	4.35	4.82	3.49	3.22	4.92	5.34	
Но	0.85	0.88	1.08	0.83	0.86	0.76	1.19	1.02	0.84	0.83	0.89	1	0.77	0.69	1.09	1.05	
Er	2.39	2.47	2.95	2.37	2.47	2.14	3.24	2.85	2.31	2.34	2.59	2.72	2.12	2.06	3.21	3.07	
Tm			0.48			0.34		0.43	0.39	-		0.44					
Yb	2.41	2.42	3.1	2.00	2.41	2.09	2.85	2.78	2.43	2.36	2.51	2.77	2.17	2.08	2.99	3.15	
Lu ⊔f	12.0	0.38	0.48	10.0	0.38	0.33	10.1	0.43	0.38	7 2	0.37	0.44	10.0	10.4	20 6	10.9	
ni To	12.0	9.90	13.0	10.0	9.00 0 95	9.74 0.89	10.1	1 01	0.97	1.3	0.75	9.01 1 NQ	12.2	10.4	20.0	10.0	
W	1.7	1.00	1.27	1.2	0.90	1,51	2.6	2.35	1.35	1.2	0.15	1.66	0.6	1.1	1.1	1.5	
	0.54		0.58			0.66	0.22	0.61	0.31			0.42	0.29		0.35	0.33	
ті	0.04																
TI Pb	12.6		16.2	9.4		15.8	12.00	19.3	12	3.40		13.7	18	12.7	11.0	15	
TI Pb Bi	12.6		16.2 0.17	9.4		15.8 0.19	12.00	19.3 0.32	12 0.16	3.40		13.7 0.17	18	12.7	11.0	15	

1 2 3	Note: Some obviously abnormal data are highlighted. The study of Hu and Gao (2008) analyzed several samples same as those analyzed in Tayl elements References:
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