



Article

# Extraction of $\gamma$ -Alumina from Low-Cost Kaolin

Khalil M. Ibrahim \* , Mohammad K. Moumani and Salsabeela K. Mohammad

Department of Earth and Environmental Sciences, Hashemite University, P.O. Box 150459, Zarqa 13115, Jordan; mmoumani92@gmail.com (M.K.M.); salsabeela1992@live.com (S.K.M.)

\* Correspondence: ibrahim@hu.edu.jo

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**Abstract:** A combined process is proposed for the utilization of local kaolin to produce alumina particles. The applied process is made in two stages: calcination at 700 °C with sodium chloride and leaching with sulfuric followed by hydrochloric acids. The optimal extraction efficiency can be obtained when the conditions are as follows: leaching temperature is at 140 °C, leaching time is 3 h 45 min and concentration of sulfuric acid is 40 wt.%. The results show that the purity of alumina reaches 79.28%, which is suitable for the production of aluminum metal. It is evident that this method of extraction of alumina from the kaolin ash is practical and feasible. The structural and morphological properties of the calcined microcrystalline powder was characterized by X-ray diffraction and scanning electron microscope (SEM).

Keywords: alumina extraction; acid leaching; kaolin; calcination

### 1. Introduction

Alumina is an important industrial mineral, which can be used as catalyst, abrasive and as adsorbent [1–4]. Alumina exists as stable form ( $\alpha$ -alumina) and in a variety of meta-stable forms including  $\gamma$ -,  $\eta$ -,  $\delta$ -,  $\theta$ -,  $\kappa$ - and  $\chi$ -alumina [5,6]. Alumina occurs naturally pure in form of the mineral corundum, although the most important natural ore for alumina is bauxite. Bauxites have been widely used in industry to produce alumina via the Bayer process [7]. Bauxite occurrences were not reported in Jordan. However, one of the alternative sources is the Al-rich kaolin, which is a hydrous aluminum silicate (Al<sub>2</sub>Si<sub>2</sub>O<sub>5</sub>(OH)<sub>4</sub>) made of the mineral kaolinite.

The development of a process for the extraction of alumina from clay was first proposed by Hoffman et al. [8]. The process consists of: (1) roasting clay at about 700 °C; (2) digesting the roasted product with dilute hydrochloric acid; (3) filtering to separate the insoluble siliceous matter from the solution containing the aluminum and soluble impurities, such as iron and alkali salts; (4) concentrating the solution; (5) precipitating the aluminum as the hydrated chloride from the concentrated solution by adding hydrochloric acid gas; (6) removing the crystals of hydrated aluminum chloride; (7) washing the crystals to remove adhering impurities; (8) calcining the hydrated chloride to obtain alumina; and (9) recovering hydrochloric acid from the waste products at the end of the process. Synthesis of alumina from kaolin were also described by Hulbert and Huff [9] by comparing acid-leaching of a calcined kaolin using nitric, sulfuric and hydrochloric acid. They concluded that the rate of alumina leaching is fastest with hydrochloric acid, slower with sulfuric acid, and slowest with nitric acid. Recent development regarding alumina production can be found in several publications [4,7,10–13]. Some examples of alumina production from kaolin comes from Nigeria and Saudi Arabia.  $\gamma$ -Al<sub>2</sub>O<sub>3</sub> was synthesized using Kankara kaolin in Nigeria [4]. The process consists of calcination at 750 °C for 2 h to prepare metakaolin, followed by leaching with 96 wt.% H<sub>2</sub>SO<sub>4</sub> and reaction with ammonium sulfate to form ammonium aluminum sulfate. Calcination at 850  $^{\circ}\text{C}$  for 3 h of ammonium aluminum sulfate was employed to obtain the  $\gamma$ -alumina [4]. The corresponding alumina extraction was about

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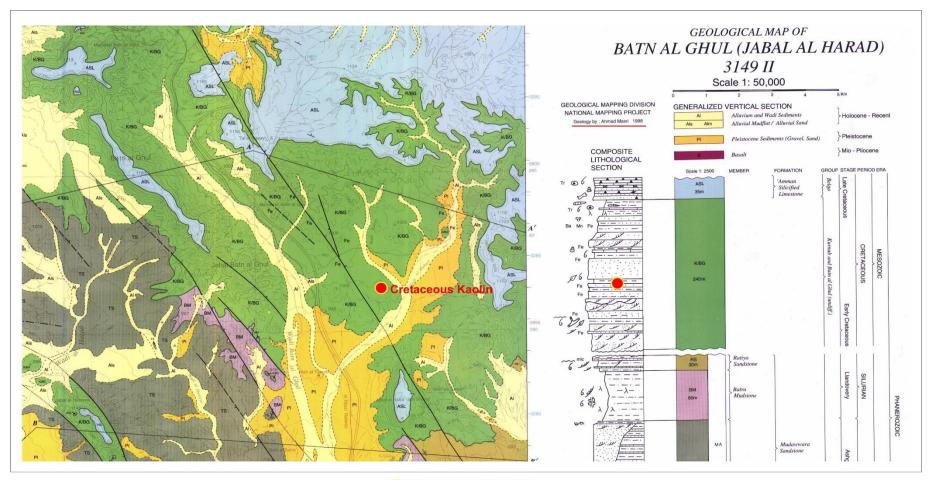
85%. Kaolin from Riyadh area in Saudi Arabia was calcined at temperatures varying between  $400\,^{\circ}\text{C}$  and  $800\,^{\circ}\text{C}$  for different calcination periods (5–120 min). The calcined kaolin was then leached with 3 M HCl acid under different leaching conditions [11]. The corresponding alumina extraction was about 63%.

Wide occurrences of kaolin are reported in Jordan [14–16], the most important of which are those occurring in south Jordan. Kaolin in south Jordan is thought as a suitable substitute from which alumina of high purity can be produced. The demand for alumina is increasing in Jordan as it is used in the ever-growing phosphoric acid and fertilizer industry, therefore increase in Jordan imports for alumina and other Al-sources are reported. Al-source is used to produce aluminum fluoride as a by-product. The objective of the paper is to assess the suitability of producing alumina by utilizing low cost raw materials such as the known huge kaolin deposits in south Jordan occurring in Harad Formation and the rock salt (sodium chloride) that occurs in huge amount in the Dead Sea and being extracted as by-product during potash harvesting. The process we applied in alumina extraction is a modified process different from these described earlier. The main goal of modification is to reduce preparation costs by utilizing low-price raw materials and enhance purity of the product by increasing the efficiency of alumina extraction.

## 2. Geological Setting

The combined name Kurnub/Batn Al Ghul (K/BG) was first used by Masri [17] and later by others [18,19] to incorporate the entire siliciclastic sequence exposed in Batn Al Ghul area underlain by Paleozoic rocks and overlain by the Campanian Amman Silicified Limestone (Figure 1). The K/BG is 240–150 m thick, generally consists of fine-, medium- to coarse-grained, varicolored, cross-bedded, moderate to well-sorted sand and sandstone interbedded with silty clay, clay, mottled clayey silt, clayey sand, siltstone and iron crusts [18,19]. It has been subdivided into two formations which are Harad and Fassu'a formations [19]. Kaolin samples of this study are Cenomanian-Turonian age [17]. They were collected from the Harad Formation, about 75 Km to southeast of Ma'an city within coordinates (29°39′3.94″ N and 35°57′20.17″ E).

Preliminary evaluation of the Harad Formation kaolin deposit indicates that the kaolin, after beneficiation process, is suitable for low price applications such as manufacturing of common bricks, medium-fired bricks and sanitary ware [20]. According to Yasin and Ghannam [16], the total inferred resource of the kaolin in the Harad Formation is about 2.2 billion ton with thickness varying from 7.5 to 47 m. The annual extraction in 2004 and 2005 was 30,000 Ton. The produced kaolin is used mainly in cement and ceramic industries and prices are always low, less than 5 US\$/ton [16].



Cretaceous Kaolin,
Batn Al Ghul Group
Harad Formation (Cenomanian-Turonian)

Figure 1. Geological map of the sampled Cretaceous kaolin deposit in south Jordan [17].

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#### 3. Material and Method

Representative kaolin sample was first cleaned from the weathering crust and carbonate by using diluted acetic acid and distilled water. The sample was then crushed and powdered and screened to particles size below 75  $\mu$ m. All of chemical used are analytical grades. Mineralogical and chemical characterization of the kaolin sample and the end products was carried out by X-ray diffraction (XRD) technique and X-ray Fluorescence (XRF). The following analytical instruments were used: Shimadzu X-ray diffractometer (MAXima XRD-7000); Shimadzu X-ray Fluorescence (Lap Center XRF-1800); and Bruker X-ray Fluorescence (S4 Pioneer). For morphological structure and identification of the powder size, samples were analyzed using SEM.

### 4. Experimental Work

A porcelain boat, containing 100 g kaolin and 37 g halite (sodium chloride), was calcined at 700 °C for 3 h and then air-dried and 119 g of kaolin ash were obtained as illustrated in the flowchart in Figure 2. The use of halite in the calcination is a slight modification from the processes described earlier by Hoffman et al. [8], Hulbert and Huff [9] and others [4,7,10-13]. The advantage of using halite is to reduce quantities and concentration of  $H_2SO_4$  compared with other works which lead to reduce cost of production and enhanced extraction efficiency.

Fifty grams of kaolin ash were leached by 200 mL of sulfuric acid solution ( $40 \text{ wt.}\% \text{ H}_2\text{SO}_4$ ) in a Pyrex round-bottom flask equipped with a reflux condenser in oil bath at constant temperature about 140 °C under constant stirring for 3 h and 45 min. Afterwards, 120 mL of HCl were added in a round-bottom flask in a hot water for 30 min. This resulted in the appearance of small amount of solution that was filtered by filter paper. Sodium hydroxide solution (30 wt.%, NaOH) was added to the filtrate until the pH value became equal to 11.3. The addition of NaOH to the process is another modification from previous works. The objective was to enhance the removal efficiency of the impurities in the end-product such as iron and titanium oxides. Following the addition of sodium hydroxide, dark reddish-brown precipitates were readily formed (Figure 3). Then, the mixture was filtered to produce the alkaline sodium aluminate solution. A reddish-brown precipitate was tested by HNO<sub>3</sub> and potassium thiocyanate (KSCN) to test for iron, and the resultant solution became deep red color (Figure 3a).

As in the flowchart in Figure 2, the alkaline sodium aluminate solution obtained in this process was bubbled with carbon dioxide gas until a white precipitation was formed (Figure 3b). The white precipitate was filtered and washed with distilled water several times until the pH value became equal to 9.1. On the next day, the pH increased to 10.9 and was rewashed with distilled water several times until the pH decreased to 10.2, and then calcined at  $600\,^{\circ}$ C for 2 h and sent for characterization (Figure 3c).

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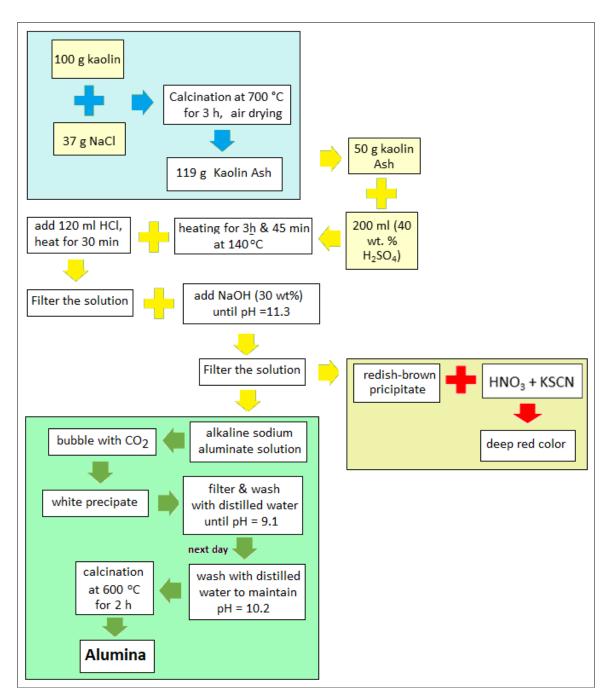
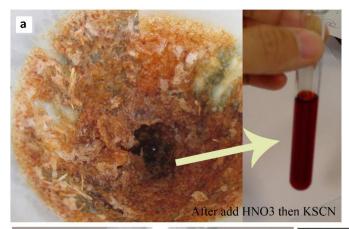


Figure 2. Flowchart summarizes steps of alumina extraction from the kaolin.

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**Figure 3.** Products formed during the experiment: (a) reddish-brown precipitate (thernadite); and (b,c) white precipitate (alumina).

### 5. Results and Discussion

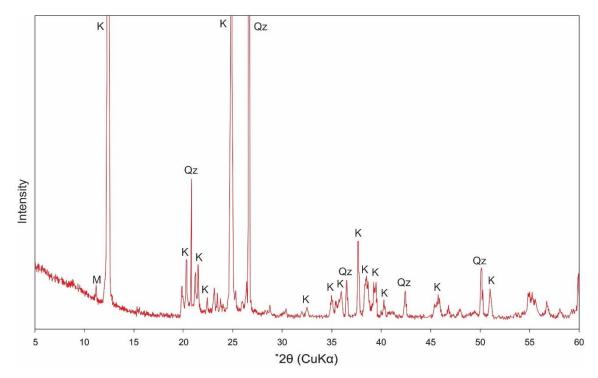
# 5.1. Original Kaolin

Chemical analysis of the original kaolin sample from the Harad Formation and the resultant calcined kaolin ash are presented in Table 1. Chemical analysis of the original kaolin indicates that it contains 53.90 wt.%  $SiO_2$  and 29.90 wt.% of  $Al_2O_3$ . Similar results were obtained by Gougazeh and Buhl [20] for  $SiO_2$  (58.02 wt.%) and for  $Al_2O_3$  (28.00%). The  $SiO_2/Al_2O_3$  ratio in the original kaolin is equal to 1.80. The  $SiO_2/Al_2O_3$  ratio in kaolinite mineral is usually 1.178 [21]. The high content of  $SiO_2$  in the sample implies the presence of free silica in the form of quartz. This is evident from the X-ray diffractogram of the sample in Figure 4. Kaolinite was the predominant mineral, followed by quartz. Traces of other minerals were indicated, such as illite-muscovite, Fe-bearing minerals (hematite), anatase and feldspar.

**Table 1.** Chemical analysis of major oxides (wt.%) before and after calcination.

Material	Fe <sub>2</sub> O <sub>3</sub>	TiO <sub>2</sub>	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	K <sub>2</sub> O	MgO	Na <sub>2</sub> O	CaO	LOI	Total
Original kaolin	1.80	1.87	53.90	29.90	0.18	0.093	0.019	0.14	12.00	99.90
Kaolin ash	1.55	1.50	44.70	23.90	0.63	0.127	16.40	0.34	10.70	99.85

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**Figure 4.** X-ray diffractogram of the kaolin sample: M, illite muscovite; K, kaolinite; and Qz, quartz.

Chemical analysis of the original kaolin sample contains variable amounts of other oxides which indicate presence of impurities as they are not related to kaolinite structure. The oxides include mainly  $Fe_2O_3$  and  $TiO_2$ . The former makes about 1.80 wt.% and the latter is 1.87 wt.%. Both oxides are probably related to the presence of hematite, ilmenite and rutile as accessory mineral phases in the kaolin deposit.

Chemical analysis of the kaolin ash (Table 1) indicates that it contains 44.70 wt.% of  $SiO_2$  and 23.90 wt. of  $Al_2O_3$  with  $SiO_2/Al_2O_3$  ratio equal to 1.87, which is similar to the ratio in the original kaolin sample.  $Na_2O$  increased from 0.02% in the original kaolin to 16.40% in the kaolin ash due to calcination with NaCl (see Figure 2).

## 5.2. Reddish-Brown Precipitate

The reddish-brown precipitate formed in the sample during the experiment (Figure 3a) was evaluated.

The average chemical composition of the red precipitate as measured from two samples indicate that it is enriched with Na<sub>2</sub>O and SO<sub>3</sub>, as illustrated in Table 2. The Na<sub>2</sub>O makes up to 54.99 wt.% and the SO<sub>3</sub> is 27.74 wt.%, with Na<sub>2</sub>O/SO<sub>3</sub> ratio equal to 1.98. Both components are the main constituents of the mineral phases thernadite and mirabilite, which have Na<sub>2</sub>O/SO<sub>3</sub> ratio equal to 0.774. Mineralogical investigation by the XRD method indicates that the red precipitate exhibits thernadite structure, as shown in Figure 5. The high Na<sub>2</sub>O/SO<sub>3</sub> ratio in the sample compared with thernadite implies presence of other phases that host the excess Na<sub>2</sub>O. Considerable amounts of SiO<sub>2</sub> and Al<sub>2</sub>O<sub>3</sub> and CaO are reported in the red precipitate. They along with the excess Na<sub>2</sub>O probably relate to the formation of sodium aluminum silicates possibly as analcime zeolites and calcium silicates in the form of plombierite (Figure 5). Formation of other combined phases cannot be ruled out.

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Material	Fe <sub>2</sub> O <sub>3</sub>	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	K <sub>2</sub> O	MgO	Na <sub>2</sub> O	SO <sub>3</sub>	CaO	Total
Red Precipitate	0.98	5.20	7.70	0.80	0.91	54.99	27.74	1.67	100.00
Alumina	0.10	0.56	79.28	-	-	19.48	0.08	0.14	99.64

Table 2. Chemical analysis of major oxides (wt.%) in the final products.

The red color resulted from the addition of HNO<sub>3</sub> and KSCN to the precipitate indicates presence of iron. The Fe<sub>2</sub>O<sub>3</sub> content in the precipitate is 0.98%.

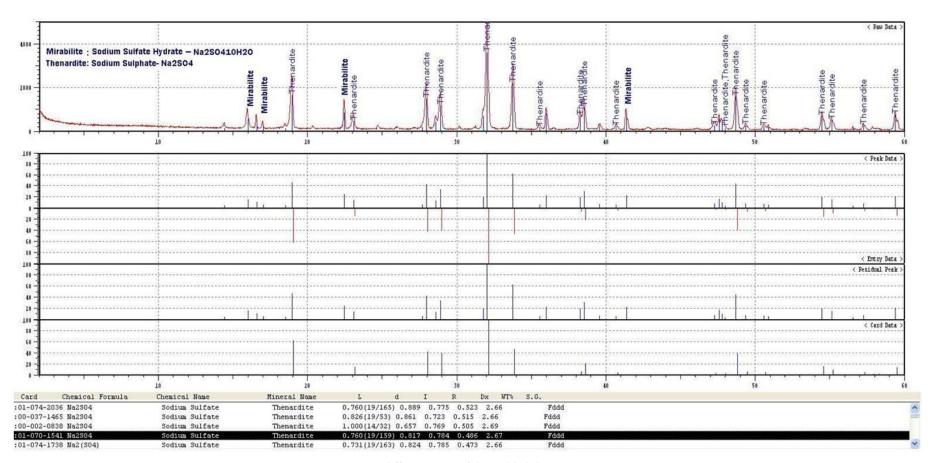
### 5.3. Alumina

The characterization of the calcined white powder using X-ray diffraction, Scanning Electron Microscope (SEM), SEM-EDX and chemical analysis by XRF indicates the presence of alumina as the major constituent. The X-ray diffractogram presented in Figure 6 indicates presence of meta-stable forms  $\gamma$ -alumina as a major constituent along with traces of sodium aluminum oxide (sodium aluminate). The formation of  $\gamma$ -alumina was supported through comparing with JCPDS 29-63 and with X-ray diffractogram given by Hosseini et al. [7].

The presence of Na-phase within the  $\gamma$ -alumina is emphasized from the XRF and SEM-EDX analyses. As listed in Table 2, the Al<sub>2</sub>O<sub>3</sub> wt.% extraction in the final product is 79.28%, which is higher than the alumina extraction (63%) observed by Al-Zahrani and Abdul-Majid [11]. The Al<sub>2</sub>O<sub>3</sub> content resembles the value obtained by Kianinia et al. [12] for alumina produced from Hamedan kaolin. The impurities in the  $\gamma$ -alumina are attributed to presence of Na<sub>2</sub>O equal to 19.48 wt.%, along with traces of SiO<sub>2</sub> (0.56%), CaO (0.14%) and Fe<sub>2</sub>O<sub>3</sub> (0.10%).

In general,  $\gamma$ -alumina can be used as catalyst or as catalyst support. This is due to its high specific surface area and thermal stability until temperature of 1030 °C. According to Belver et al. [22], thermal stability of  $\gamma$ -alumina can be improved in the presence of some additives including CaO and SiO<sub>2</sub>. Therefore, it can be concluded that the presence of such impurities in the prepared alumina will enhance its thermal stability.

The SEM micrograph in Figure 7 shows that the alumina grains have flack shape and are uniform in size. Most of the grains are less than 10  $\mu$ m, but some grains may reach up to 40  $\mu$ m. The grains seem larger than those synthesized by Hosseini et al. [7] through reaction of meta-kaolin with  $H_2SO_4$  solution using direct precipitation in ethanol. Two spot SEM-EDX analysis show that  $Al_2O_3$  is more than 90% and  $Na_2O$  is less than 10% (Figure 7)



**Figure 5.** X-ray diffractogram of the reddish-brown precipitate.

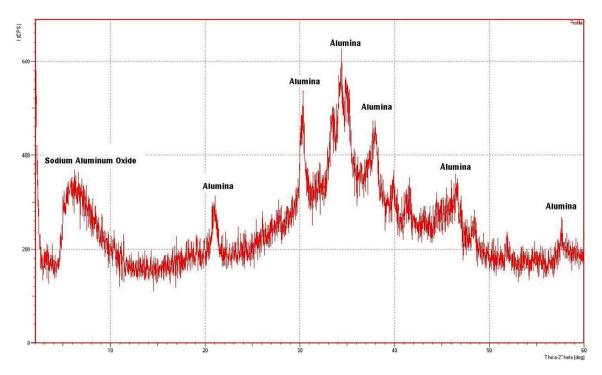


Figure 6. X-ray diffractogram of the extracted alumina.

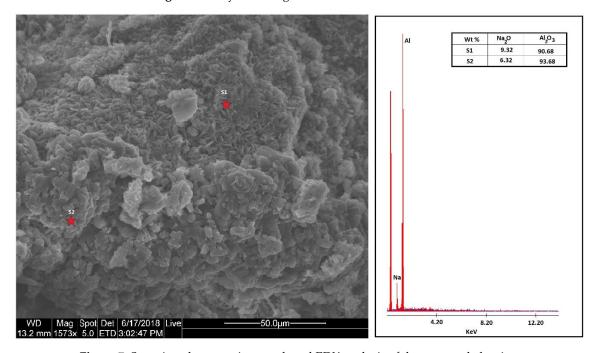


Figure 7. Scanning electron micrograph and EDX analysis of the extracted alumina.

## 6. Conclusions

The overall objective of this study was to propose and develop a process for the extraction of alumina from the kaolin deposits that occur in huge quantities in south Jordan. The process we applied in alumina extraction is a modified process different from those described earlier. The modification reduced preparation costs by utilizing low price raw materials and increased alumina extraction efficiency.

The extraction process comprised of calcination with NaCl followed by two stages of leaching with acids. The alumina extraction of kaolinite ash was optimally synthesized at the experimental conditions of: 200 mL  $H_2SO_4$  solution (40 wt.%), 140 °C, 3 h 45 min reaction time, 120 mL HCl for 0.5 h,

and NaOH solution (30 wt.%,). The process leads to synthesis of alumina powder from the kaolin ash. The purity of  $Al_2O_3$  was 79.28 wt.%, which is suitable for production of aluminum metal or to produce aluminum fluoride as a by-product from phosphoric acid production. The alumina impurities are attributed mainly to presence of sodium aluminate.

**Author Contributions:** K.M.I. contributed the literature search, supervised the experimental work and chemical analysis, participated in data interpretation and wrote the paper; M.K.M. and S.K.M. contributed the literature search, designed and performed the experiments, performed the chemical analysis and participated in data interpretation.

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**Conflicts of Interest:** The authors declare no conflict of interest.

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