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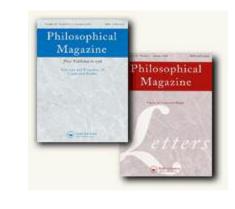
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Hardening and thermal stability of nanocrystalline AlMg4.8 powder

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Abstract

Ball milled nanocrystalline AlMg4.8 powder was investigated in terms of hardening and thermal stability. The validity of the Hall-Petch relation was confirmed down to the minimum grain size of ~ 44 nm. Prolonged milling in the range of the minimum grain size still increased the hardness. This development was discussed in terms of contamination effects and the influence of full and partial dislocations. Concerning thermal stability, recovery processes occur in the range of 100 – 230 °C while substantial grain growth starts at a temperature of ~ 250 °C. The enthalpy release for recovery was detected to be ~ 39 J/mol and ~ 208 J/mol for grain growth. Dynamic strain aging was indicated by an activation energy for recovery of Q ~ 120 kJ/mol. The activation energy of grain growth was calculated by means of the Kissinger theory (Q = 200 - 210 kJ/mol) and using the results of static grain growth (Q = 204 kJ/mol).

Keywords: Aluminium alloys, Ball milling, Nanocrystalline materials, Differential scanning calorimetry, Hall-Petch

1. Introduction

Nanocrystalline (NC) materials feature grain sizes smaller than 100 nm. Gleiter [1] was one of the first who reported the outstanding mechanical and physical properties of this group of materials in contrast to conventional coarse-grained materials. This behaviour is due to the extremely high volume fraction of grain boundaries. The excess Gibbs free energy stored in the two-dimensional defects is the reason for a thermodynamically non-equilibrium state. The stability of NC materials with respect to grain growth is often discussed in literature [2]. When NC materials are investigated, the synthesis process always has to be considered, as each route introduces specific contaminations. Impurities and inhomogeneities can strongly alter the mechanical behaviour and especially the thermal stability [3]. The enhanced resistance to grain growth is associated to compositional parameters as well as to microstructural inhomogeneities as vacancies [4]. NC

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materials can be synthesized by different methods for instance by inert gas condensation, pulsed electro deposition, chemical and physical vapour deposition and ball milling followed by a consolidation process. The latter is applied in the present study due to the high versatility of the ball milling process yielding very low grain size values in a huge variety of crystalline materials [5]. But synthesis of nanostructured materials via powder metallurgy comprising ball milling can introduce a high quantity of impurities. One major source is the attrition of the milling tools; the other is the incorporation of gases due to the generation of activated surfaces in the milled powder. Therefore these factors have to be thoroughly monitored regarding the evaluation of the mechanical properties.

The validity of the Hall-Petch relationship, describing the hardness/strength increase as a function of grain size is still under discussion [6, 7]. In recent years it was found that this equation holds for polycrystalline materials in a wide range concerning their grain size. In fact for materials with very small grains, in the range of less than approximately 100 nm, its validity is not clear as the deformation mechanisms change and contamination effects increase. Mechanisms like grain-boundary sliding, grain boundary diffusion and mechanical twinning are supposed to dominate, rather than dislocation based plasticity. Twinning is associated with the emission of partial dislocations, which are favoured from an energetic point of view. Thus a breakdown of the relationship and even an inversion of this relationship have been discussed [8-11]. In contrast, dislocation pile-up was observed recently during TEM in situ straining by Youssef et al. [12] in NC Cu. A finding if the Hall-Petch relation is still valid for the NC regime would also elucidate some questions concerning the character of thermal activated processes.

The present study addresses the investigation of the microstructural evolution and the change in the composition during milling of the alloy AlMg4.8. The thermal stability of the milled NC powder is studied quantitatively by means of DSC measurements and grain size evolution detected by XRD and TEM respectively. The results concerning microstructural changes during synthesis and annealing are related to the micro-hardness. The sensitivity of the mechanical properties of NC materials concerning impurities was also evaluated.

2. Experimental procedure

The binary alloy powder AlMg4.8 was synthesised by inert gas atomisation carried out by ECKA Granules. The powder showed an average particle size of 50 μ m. The chemical composition was determined by inductively coupled plasma (ICP) measurement. The gas content was investigated by means of hot gas extraction.

Milling was performed in the planetary ball mill "pulverisette6" from Fritsch using hardened steel vial and balls. The ball to powder weight ratio was 5:1, whereas no process control agent was added. To reduce contaminations during milling all experiments were carried out under Argon environment. The milling times given in the present study are the total of 15 min milling steps. The incremental procedure of 15 min milling and a phase of 30 min cooling avoided strong temperature increase. The powder output is around 6 wt. % lower than the input which is mainly attributed due to the adherence of the ground material on the milling balls.

All investigations concerning the thermal stability were carried out on the powder AlMg4.8 milled for 8 h. Annealing experiments of 0.5, 3 and 100 h duration were accomplished in the temperature range 150°C - 400°C with increments of 50 °C. Subsequently, Vickers micro-hardness HV 0.05 was determined using a LECO LM300AT micro-hardness tester. The powder particles were preliminarily fixed by resin vacuum infiltration and subsequently grinded and polished before

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testing the core hardness. The given hardness values represent the average of 15 separate measurements. A hardness scatter of ~ 12 HV and a grain size scatter of ~ 7 nm at a grain size of 45 nm indicate a good homogeneity of the milled material. This proves the reliability of microhardness testing of the coarsened powder particles. The particle size is ~ 500 μ m in the asmilled state due to cold welding during milling. Calorimetric studies on the as-milled powder after 8 h milling time and subsequently annealed powders were carried out by means of a differential scanning calorimeter DSC 204/F1 Phoenix from Netzsch. The powders were sealed in Al pans and investigated at a heating rate of 10 K/min in flowing nitrogen atmosphere. Also heating rates of 5 and 20 K/min were applied in the as-milled state. The DSC was calibrated for the required particular heating rates before starting the measurement.

Transmission electron microscopy (TEM) was performed on a Philips CM 200 microscope. As the powder particle size increased during milling, grinding and subsequent electrochemical polishing were applied for preparation. Selected area diffraction pattern were taken. The sample area contributing to the diffraction patterns was kept constant at 25.3 μ m². The quantification of the crystallite size and the dislocation density was done with X-ray diffraction (XRD) experiments. The experiments were performed with a double crystal diffractometer setup, according to Wilkens and Eckert [13], corrected for instrumental broadening. In this case up to five diffraction profiles were recorded for each sample, from the (111) to the (222) reflection. The Multiple Whole Profile (MWP) fit procedure, developed by Ungár et al. [14] and implemented into a software package by Ribárik et al. [15] was used for profile analysis. In this procedure, simulated profiles are fitted to the recorded profiles. This is done simultaneously with ab initio theoretical functions for the strain and size induced profile broadening. As a result values for the median and the variance of a log normal crystallite size distribution are received. For clarity these values are converted into volume averages according to Hinds [16]. During recent years this method became well established for investigating nanocrystalline and ultrafine grained materials [17]. In Fig. 1 the massive peak broadening due to the nanosized grain structure in the as-milled state is shown. After heat treatment at a temperature of 400 °C for 3 h, the peak becomes narrower as grain coarsening and dislocation annihilation occurred. These peaks indicate the increase of the average grain size from 73 nm to 297 nm.

[Insert figure 1 about here]

Fig. 1. Comparison of the peak widths of AlMg4.8 in the as-milled (8 h) and annealed (400 $^{\circ}$ C3 h) state obtained by XRD.

3. Results

3.1. Synthesis

3.1.1 Microstructure and hardness evolution during synthesis. The microstructure in the asatomized state is dendritic. The microstructure of the milled powder was investigated by TEM after 8 h of milling (Fig. 2). According to XRD analysis the as-milled specimen is a supersaturated fcc solid solution. The TEM bright field image of the powder depicts a grain structure in the range of 50 - 120 nm. The grain size distribution is homogenous and monomodal and the outlines of the grains are diffuse. The grain size obtained after 8, 32 and 50 h milling time was determined by means of XRD and was found to be 73, 45 and 44 nm respectively (Fig. 3). No information is

given by XRD investigation about the fraction of large and small angle grain boundaries. The results of these two characterisation techniques are in good agreement, indicating that the area investigated by TEM is representative.

[Insert figure 2 about here] Fig. 2. TEM bright field image of the as-milled nc AlMg4.8 powder and selected area diffraction pattern.

The micro-hardness increased with the milling time which is shown in Fig. 3. No hardness plateau was reached within the first 50 h of milling. According to the hardness values, AlMg4.8 still shows hardening even if a nano-scaled microstructure was adjusted. This result differs strongly from the findings of Koch [18]. In his study Zn show no steady increase of hardness in the first hours of milling. Nevertheless this strongly depends on the milling parameter and the powder. The rapid decrease in grain size within the first hours of milling and its steady state after prolonged milling is consistent with the findings of Shaw et al. [19]. This implies the fast generation of dislocation networks and the generation of small grains [18]. The misorientation of the grains also increases with increasing milling time [20]. This probably leads to the found increase in hardness.

[Insert figure 3 about here]

Fig. 3. Change of micro-hardness and crystallite size as a function of milling time.

3.1.2 Composition and contamination during synthesis. In order to evaluate the contaminations introduced in the powder by milling, the composition of the as-sprayed powder and as-milled powder after a milling time of 32 h was investigated (Table 1.). The element iron (Fe) is the only contamination which was introduced by the milling tools during service. Its content increased from 0.070 wt% up to 0.078 wt%. Nevertheless, the Fe contamination is still on a very low level. The other main source of contamination is the absorption of gases as hydrogen (H), nitrogen (N) and oxygen (O) during the milling process. This could also happen if the process is performed in argon atmosphere due to adhering molecules on the particle surfaces. The relevant processes during milling as cold welding, fracturing and deformation generate activated surfaces. This promotes intense diffusion of the H, N and O atoms into the material.

[Insert table 1 about here]

The values determined by hot gas extraction in the as-atomized state and after milling durations of 8 h and 32 h are given in Table 2. The high oxygen content could be justified by the reaction with Aluminium to Al_xO_y . The rather slow increase of H and N content could be deduced from the lack of reaction with the milled material. Thus, these elements are interstitially solved in the AlMg4.8 lattice. The low contents especially of H and N are in good accordance with the low solubility of these elements as found in the phase diagrams [21].

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[Insert table 2 about here]

3.2. Thermal stability

3.2.1 Grain size and hardness evolution. In Fig. 4 the crystallite size and the dislocation density evolution are given in dependence of the annealing temperature. At an annealing time of 3 h for temperatures larger than 250 °C the average grain size increased steadily. Concurrently the dislocation density decreased. After annealing at 400 °C the dislocation density is below 10^{13} m⁻², which is the detection limit of the applied XRD based method.

[Insert figure 4 about here]

Fig. 4. Crystallite size and dislocation density evolution in dependence of annealing for 3 h at different temperatures of AlMg4.8 preliminary milled for 8 h.

Fig. 5 shows the hardness evolution of the powder milled for 8 h at different annealing times and temperatures. The plotted points and error bars represent the average value of 15 hardness measurements with the corresponding standard deviation. A slight hardness maximum is found at a temperature of 100 °C for an annealing time of 3h and 100 h. This peak is shifted to around 150 °C at the shorter annealing time of 0.5 h. The hardness then steadily decreases for all annealing times at higher temperatures at a slowly decreasing slope. As annihilation and polygonisation are time-dependent the hardness at a given temperature above 150 °C is higher at shorter annealing times.

[Insert figure 5 about here] Fig. 5. Microhardness of AlMg4.8 powder milled for 8 h after annealing at different temperatures for 0.5 h, 3 h and 100 h.

3.2.2 DSC measurements. DSC measurements were performed on the as-milled powder (8 h) and on annealed samples with different heating rates. The curves show qualitatively similar characteristics at different heating rates. Therefore, in Fig. 6 DSC measurements after different annealing temperatures at an annealing time of 3 h are shown exemplarily, where a heating rate of 10 K/min was used.

[Insert figure 6 about here] Fig. 6. DSC curves for the as-milled and annealed sample at a heating rate of 10 K/min.

The calorimetric data on the NC powders in the as-milled state show two pronounced exothermic peaks. A low temperature peak (LT) is found in a temperature range of 100 - 230 °C. A high temperature peak (HT) starts at ~ 260 °C and is more pronounced than the LT-peak. After annealing at 200 °C the LT-peak disappeared and the height of the HT-peak slightly decreased. After annealing at even higher temperatures up-to 350 °C the height and the slope of the HT-peak continuously decreased.

As expected the peaks become more pronounced with increasing heating rate and the peak temperatures were shifted to higher values when the heating rate was increased. The respective peak temperatures for all applied heating rates are given in table 3.

[Insert table 3 about here]

3.2.3 Activation energy of grain growth. The DSC measurements presented in Fig. 6 show the two exothermic peaks, designated as low (LT) and high (HT) peak in the as-milled state. The LT-peak refers to the relaxation of thermodynamically non-equilibrium grain boundaries and recovery processes. The HT-peak refers to the excess of residual Gibbs free energy of grain growth processes. In this study grain growth was confirmed by TEM investigation of the samples at 200 °C and 400 °C. These temperatures are before and after the HT peak respectively. Tschöpe et al. [22] showed for NC Pt that the Gibbs free energy can be estimated as the enthalpy change. The entropy contribution in this regard is negligible small.

In order to evaluate the activation energy of these two processes quantitatively, two different methods were pursued. One is based upon the shift of the DSC peaks caused by the variation of the heating rate also known as Kissinger analysis. The other rests upon the grain size evolution during annealing. Both calculations are given in the Appendix. Table 4 shows the results from the Kissinger analysis from which an activation energy Q of 120 - 126 kJ/mol was obtained for the LT-peak and 201 - 210 kJ/mol for the HT-peak. Table 4 shows that the exponent used in the Kissinger approach has only a minor influence on the results.

[Insert table 4 about here]

With the static grain growth approach an activation energy of 204 K/J/mol was obtained for the HT peak. This value is in the range of the results calculated by the Kissinger analysis and indicates a good agreement between these methods.

3.2.4 Enthalpy release. The area below the exothermic peaks indicates the amount of the enthalpy release. In the following a quantitative approach was pursued to determine the enthalpy release of NC AlMg4.8 based on the DSC plot of the as-milled (8 h) powder measured at a heating rate of 10 K/min. The enthalpy releases of the exothermic LT and HT reactions were calculated by integrating the DSC signal. As the used DSC apparatus allows only measurements up to 400 °C the HT-peak could not be fully recorded. Therefore the area below the peak was extrapolated by integrating the area from the onset of the peak up to the maximum and multiplied by the factor 2. This assumption of a symmetric DSC peak shape of NC materials was reported by Eckert et al

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[23]. The heat release of the LT- and HT-peak were calculated to be 39 J/mol and 208 J/mol respectively.

3.2.5 Hardening in the NC regime and impurity influence. First the impurity effect introduced during the process on the mechanical properties is discussed. The increasing contamination content probably only has a minor influence on the rising hardness with longer milling times. The amount of the Fe content after 32 h milling is negligible small. Under the precondition that all oxygen atoms reacted with Al, around 180 ppm and 480 ppm Al_2O_3 were created in AlMg4.8 after 8 h and 32 h milling time, respectively. Summarizing these results a maximum oxide content of ~ 0.2 wt. % and ~ 0.5 wt. % can be assumed for 8 h and 32 h milling. On the other hand the hardness increases from 153 to 212 HV0.05 for powders milled for 8 or 32 hours. Thus the small increase in the second phase content probably only has a minor influence on the mechanical properties.

Therefore, the hardness increase has to be correlated to microstructural changes. Concerning the grain size, a decrease from 73 nm after 8 h milling down to 45 nm after 32 h and 44 nm after 50 h milling was found. The small grain size almost excludes conventional hardening mechanism as complete dislocation pile ups. Nevertheless, full dislocations are necessary for grain refinement. As this happened here up to 32 h milling time conventional dislocation multiplication mechanisms might still be active. However longer milling up to 50 h yielded only negligible grain refinement indicating the lower grain size limit of the current alloy. Proceeding hardness increase after milling times > 32 h might be the result of full or partial dislocation multiplication associated with mechanical twinning [24]. Polygonisation came to an end at the mentioned minimum grain size of ~ 44 nm.

An estimation concerning the validity of the Hall-Petch relation (1) is given in Fig. 7.

$$H_M = H_0 + k \frac{1}{\sqrt{d}} \qquad (1)$$

Where H_M is the hardness after a specific milling time, H_0 is the hardness after long time annealing, k is the Hall-Petch constant and d the average grain diameter. H_0 was determined after annealing at 530°C for 3 h to be 48 HV0.05.

[Insert figure 7 about here]

Fig. 7. Correlation of grain size and hardness of AlMg4.8 linearised in a Hall-Petch plot after different process routes.

A slope of 0.52 is obtained which stands for the exponent in a Hall-Petch type equation. This verifies the applicability of the Hall-Petch relation down to a grain size of 45 nm. This is consistent with the conclusions in the review of Dao et al. [25]. In this range the hardness increase probably is due to full dislocation pile-ups indicated by the consistence with the linear Hall-Petch curve. At a grain size below 45 nm the hardness increase proceeds at a higher rate. Even if deformation mechanisms like partial dislocations and twins are active and present in this grain size regime, the hardness increase can not be exclusively dedicated to this phenomenon. A proceeding dispersion of the contamination and even a slight increase in the contamination content with ongoing milling could also influence the hardness increase.

4. Discussion

4.1 Thermally activated processes

The gained information about recovery and grain growth reactions confirms the hardness evolution in dependence of the annealing time (Fig. 5). The onsets of the hardness decrease after 3 h and 100 h annealing start between 100 °C and 150 °C. At 0.5 h annealing time the onset is shifted to higher temperatures above 150 °C. The first exothermic peak (LT) is found in the DSC plot at around 160 °C. This implies that the microstructure is more stable in the DSC tests than in hardness measurement. This finding is comprehensible as the gradual temperature increase displaces the peak to higher temperatures.

The assumption that the LT-peak refers to recovery processes is confirmed by the XRD investigations shown in Fig. 4. The dislocation density strongly decreased during annealing at a temperature around 250 °C which can be deduced to annihilation, polygonisation and relaxation of non equilibrium grain boundaries. The decrease in heat release and the disappearance of the peak with increasing annealing temperature as shown in Fig. 6 also confirms this assumption. It indicates that the energy stored by the dislocations decreased when a preliminary heat treatment was conducted. As the amount of energy per dislocation is estimated to be constant, this is originated by a decreasing number of dislocations.

The HT-peak which is supposed to refer to grain growth and recovery processes along high angle grain boundaries shows a decrease in the size and also a slight decrease in the slope if the asmilled and the annealed state is compared. This means that a small amount of energy was released during preliminary annealing setting the grain boundaries in a state of higher thermodynamic stability. These changes in the character of the non-equilibrium grain boundaries in the as-milled state are mainly attributed to reorganisation and relaxation processes. The slope as well as the height of the peak stayed constant until an annealing temperature of around 250 °C. Therefore only slight grain coarsening occurred in the temperature range < 250 °C, comparing the grain size of 73 nm in the as-milled state and 146 nm after 3 h annealing at 250 °C (Fig. 4). According to Fig. 5, the slope of the hardness decrease related to recovery is higher than that related to grain growth. Thus, if the dislocation density is strongly reduced by annihilation, the hardness could be much lower even if the grain size has not varied significantly. This underlines the meaning of dislocations for NC materials and points out that the recovery rate is an important factor regarding the mechanical properties. The high dislocation density of the AlMg4.8 powder milled for 8 h indicates that the grain refinement has not been finished yet. This is confirmed as the grain size decreased with longer milling time as already mentioned. Even after 32 h milling the final microstructure was not obtained. This value is given by the competition between mechanically induced disordering and thermally activated reordering processes.

4.2 Effect of the grain size on the growth exponent

Summarizing the results concerning the activation energy for the temperature range above 260 °C, this temperature refers to the onset temperature of the HT-peak in the as-milled state. With the Kissinger approach an activation energy of 201 - 210 kJ/mol, whereas the static grain growth analysis resulted an activation energy of 204 kJ/mol. The very good consistency of these results also confirms the accuracy of the used exponent n = 0.08 in the analysis of static grain growth taken from literature [26]. The activation energy measured by Wang et al. [27] on non-

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recrystallised UFG AlMg3 in the temperature range below 548 K prepared by ECAP was reported to be only 30 kJ/mol. This value is significantly lower than the result of the present study. The massive deviation can be attributed to the exponent n, where 0.5 is used for static grain growth in conventional materials as derived from Burke [28]. However, the validity of this equation is focused on conventional coarse grained materials. The exponent n has to be adapted to UFG and NC materials. The value n = 0.08 adopted in this study was empirically derived and mainly depends on the grain size and not on the material [30]. For clarification, if n = 0.5 is used in the present study, an activation energy of ~ 36 kJ/mol would be derived. This is the same order of magnitude of the result of Wang et al. [27]. The slightly higher activation energy could be explained by the higher Mg content and a smaller initial grain size.

4.3 Effect of the alloying element Mg on the activation energy

The activation energy for grain boundary migration in a solid solution is in the range of selfdiffusion or diffusion of solutes [29]. Activation energies of 144.4 kJ/mol for lattice self diffusion and 86 kJ/mol for grain boundary diffusion in Al are reported in literature [30]. The lattice self diffusion is therefore ~ 30 % lower than the calculated activation energy of 201 - 210 kJ/mol. This indicates that the activation energy is also influenced by other effects. The retardant force due to solute drag or Zener pinning is supposed to be the fundamental mechanism. Due to the high interfacial area in NC materials the particles or solute atoms can be dispersed finely and impede grain boundary motion. This grain size dependence of the pinning forces was first discussed by Michels et al. [31]. They adapted the assumption given by Burke [28] to NC materials. For the studied alloy AlMg4.8 second phase particles as Mg₁₇Al₁₂ precipitates and Al₂O₃ are supposed to be not as significant for microstructural stabilisation due to their low fraction as solute atoms. Solute Mg atoms and vacancies should have a much higher effect due to a higher quantity and a more homogenous dispersion.

May et al. [32] obtained an activation energy of 70 - 90 kJ/mol in a similar temperature range of 293 - 473 K by strain rate jump tests on UFG Al99.5. The reason for the ~ 35% lower value compared to the data presented here could be explained by the effect of the alloying element Mg. The activation energy for Mg diffusion in Al was found to be 120.5 kJ/mol [33]. Atodiresei et al. [34] obtained the activation energy for Mg diffusion in AlMg4 by mechanical spectroscopy in a comparable temperature range to be ~ 121.2 kJ/mol. This means that dislocation movement is controlled by Mg diffusion. The process is also known as dynamic strain aging or Portevin-Le Châtelier effect (PLC). The hardness evolution confirms this hypothesis as the hardness increased at around 100°C - 150°C at an annealing time of 3 h and 100 h and at 150 °C for 0.5 h. During annealing, Mg atoms can diffuse to dislocations during the process of solute redistribution and are trapped in the dilatation zone. Consequently the resistance to dislocation motion increased.

4.4 Energy stored in NC AlMg4.8

The obtained values of 39 J/mol and 208 J/mol for recovery and grain growth respectively are very high. As discussed in the chapter concerning strain hardening, the low stacking fault energy could be an important precondition for the storage of a large amount of energy inside the material in terms of dislocations.

This assumption was verified in the following by a theoretical approach, calculating the energy which is stored in the dislocation. Herein the stored energy due to dislocations is assumed to be

proportional to the dislocation density. For this approach the dislocation density of $\rho = 1.58 \times 10^{16} \text{ m}^{-2}$ shown in Fig 4 was applied in equation 2.

 $E_D = \rho E_{Dis} \qquad (2)$

Where E_D is the total strain energy of dislocations and E_{Dis} the strain energy associated with unit length of dislocation line. E_{Dis} is calculated according equation 3 [35].

$$E_{Dis} = \frac{Gb^2(\frac{1-\nu}{2})}{4\pi(1-\nu)}\ln(\frac{\alpha R}{2b})$$
 (3)

Where G is the shear modulus, b the Burgers vector, v the Poisson ratio, R is the separation between screw and edge dislocations and α is the core unity factor. Adopting G = 25.4 GPa, b = 0.286 nm, v = 0.3, $\rho = 1.58 \times 10^{16} \text{ m}^{-2}$, R = $\rho^{-0.5}$ and $\alpha = 1$, E_D was found to be ~ 213 J/mol. Considering the error due to approximations in the applied equations this finding is in good agreement with the heat release measured by DSC peak integration (Tab. 4). Consequently the emitted energy obtained from the DSC measurements can be completely attributed to dislocation recovery processes. The much higher energies detected by DSC measurements reported by Zhou et al [29] for a NC Al-Mg alloy can not be confirmed. Therein the gap between the DSC results and the calculated energy stored by dislocations were explained by other sources like twinning and non-equilibrium grain boundaries. The influence of these mechanisms concerning the stored energy seems not to be dominant for the investigated NC AlMg4.8. The different milling processes can be an explanation for this, as cryomilling in an attritor was applied. This process is known to be more effective in nanostructuring as the energy density of the apparatus is higher and recovery is suppressed due to the low temperature.

Molecular dynamic simulations on NC materials predicted that samples consisting of high angle grain boundaries (HAGB) have most of their excess energy in the grain boundaries [36]. Those consisting of low angle grain boundaries (LAGB) store the energy in the interior as elastic distortion. The small LT-peak therefore belongs to the energy stored inside the grains or inside LAGB. The enthalpy release requires lower temperatures and activation energies. The release of the energy stored inside the grain boundaries of HAGB samples needed higher activation energy. This is due to their metastable state which is generated by forming grains with high misorientation out of the dislocation networks. Solute atomes as Mg promote this stability. The existence of the LT-peak indicates that the minimum grain size has not been reached yet after 8 h milling as there are still dislocations and subgrains in the interior. This finding also explains the validity of the Hall-Petch equation as dislocation pile-ups can still occur in this condition. This assumption is confirmed when the DSC curves of AlMg4.8 milled for 8 h and 32 h are compared. Due to the longer milling time the grain size decreased from 73 nm to 52 nm. Concurrently the energy in the grain interior decreased and is located mostly in the HAGBs. Consequently the DSC curve for the milling time of 32 h did not show a pronounced LT-peak. This is also consistent with the assumption made with regard to the hardening. It was supposed that when the minimum grain size is reached full dislocation multiplication come to an end. Therefore no recovery processes in terms of annihilation and polygonisation of dislocations can be detected by DSC.

4.5 Validation of the Hall-Petch relation in the nanocrystalline regime

The results indicated that the Hall-Petch relation is valid down to the minimum grain size of ~ 44 nm as the exponent of the corresponding equation was calculated to be 0.52. The constant k

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of the Hall-Petch relation was also calculated. This result fits well with the linear function of k vs. Mg content shown in Fig. 8.

[Insert figure 8 about here]

Fig. 8. Hall Petch constant k depends on the Mg content; k value of 4.8 wt% was calculated in the present work while the other results are reported in literature [37-39].

The lower grain size limit of Al-Mg alloys is influenced by the stacking fault energy. Mohamed [40] found a correlation for ball milled material which has been confirmed later for other NC materials prepared by high pressure torsion [41]. It is known from literature that an increasing content of Mg strongly decreases the stacking fault energy of Aluminium. The recovery rate is significantly lower as cross slipping of dislocations is more difficult for metals with lower stacking fault energy due to the higher width of the stacking fault ribbon. This results in a higher level of equilibrium concerning the dislocation density at a given temperature. According to the Hall-Petch equation hardness is influenced by the grain size "d" and the parameter "k". The latter seems to depend on the equilibrium of the dislocation density which explains the observed correlation between the Mg content in Al and the parameter k. Therefore the lower grain size limit of 44 nm determined here for AlMg4.8 is lower than that of pure Aluminium [42/43].

Milling after the lower grain size limit is reached yielded further hardening and could not be explained by the Hall-Petch relationship. The description of the hardness evolution in this regime needs further investigation. This concerns also new deformation mechanisms as partial dislocation emission and absorption by grain boundaries. These mechanisms were originally predicted using molecular dynamic simulations by Yamakov et al. [44] and experimentally first observed by Chen et al. [45]. The mentioned lower stacking fault energy due to Mg also promotes the generation of partial dislocations and twins. Especially in the state where conventional dislocation multiplication already came to an end these mechanisms might be an explanation for the proceeding hardening at a stable grain size. No inverse Hall-Petch effect was found.

5. Conclusion

- AlMg4.8 was investigated concerning the correlation of grain size and hardness down to the nanometer regime. The hardness evolution was found to be consistent with the Hall-Petch relation down to a grain size of ~ 45 nm. Prolonged milling has no influence on further grain refinement. The influences of ball-milling related impurities were quantified. Even if their amount is rather small, an effect on the hardening at the minimum grain size can not be neglected. Nevertheless deformation mechanisms which are not common in conventional grained aluminium alloys as partial dislocations and twins could be active too.

- Two peaks were found in DSC measurements on as-milled and annealded samples. The activation energies Q of the thermal activated processes were derived by means of DSC analysis. The activation energy for the low temperature recovery peak was found to be 123 kJ/mol. A similar value of 121 kJ/mol was reported in literature for Mg diffusion in Al. Recovery seems to be controlled by Mg diffusion. As the investigated specimen has not reached the minimum grain size after 8 h milling, conventional dislocation mediated mechanisms still are active, which are necessary for further grain refinement. The activation energy for recovery confirms this

assumption. For grain boundary movement the average activation enery was calculated via static grain growth and Kissinger analysis to be ~ 205 kJ/mol. For Al self diffusion Q is 144.4 KJ/mol which is ~ 40% lower than the calculated values. This increase of activation energy for grain growth is explained by the large interfacial area in the material enabling effective pinning by solute atoms and vacancies. This also explains the lower thermal stability of pure Al.

- Enthalpy release Δ H during calorimetry was found to be 39 J/mol and 208 J/mol for recovery and grain growth. The energy Δ H stored by the dislocations was calculated to be ~ 213 J/mol using the measured dislocation density. Considering the approximations for the peak integration and the theoretical calculation a good consistency is found. The enthalpy release of the LT and HT-peak was attributed to the grain interior and to high angle grain boundaries respectively. This is consistent with the results of molecular dynamic simulations.

Appendix A: Calculation of the activation energy

Activation energy via Kissinger analysis. The activation energies of recovery and recrystallisation were determined via DSC measurements at different heating rates according to the Kissinger analysis [48]. The DSC peak positions (Fig. 6) in combination with the corresponding heating rate give information about the activation energy of the thermal activated processes. Using the following equation the activation energy Q of the current processes could be calculated.

$$\ln(\alpha) = -\frac{Q}{k_B T_P} + C \tag{A 1}$$

Here k_B is the Boltzmann constant and C is an integration constant. α is a function of the peak temperature TP and the applied heating rate L.

 $\alpha = L/Tn$ (A 2)

Kissinger [46] and Chen et al. [47] assumed different exponents in this equation for α . The values for the LT and HT-peaks at each heating rate were put in a Kissinger plot (Fig. 9). Based on equation (A 1) and the given equations for α the Q values could be calculated on the basis of the slopes of the linear curves which are shown in the Kissinger plot.

[Insert figure 9 about here]

Fig. 9. Kissinger plot indicating the activation energy for grain growth.

Activation energy via static grain growth. The relationship between the average grain size and the annealing time for static grain growth could be described in the form given in equation A 3 [28].

 $d^{\frac{1}{n}} - d^{\frac{1}{n}}_{0} = ct$ (A 3)

Where d is the grain size after annealing, d_0 is the initial grain size and t is the time. The exponent $n \sim 0.08$ was used by Zhou et al. [26] for ball milled NC Al for the temperature range < 450°C. It was generated by applying a nonlinear, general fitting routine to the curve of the grain size and

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time data. C is a temperature-dependent constant which could be expressed by an Arrhenius type of equation in the form of:

 $c = Ae^{(-Q/RT)} \qquad (A 4)$

Where Q is the activation energy for grain growth, R the gas constant and T is the isothermal annealing temperature. The activation energy Q can be determined from the linear plot of ln(c) and 1/T as shown in Fig. 10.

[Insert figure 10 about here]

Fig. 10. Arrhenius plot of rate constant values obtained by the growth results.

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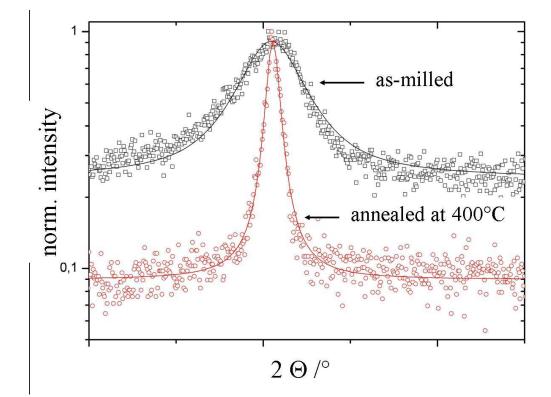
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Figure Caption

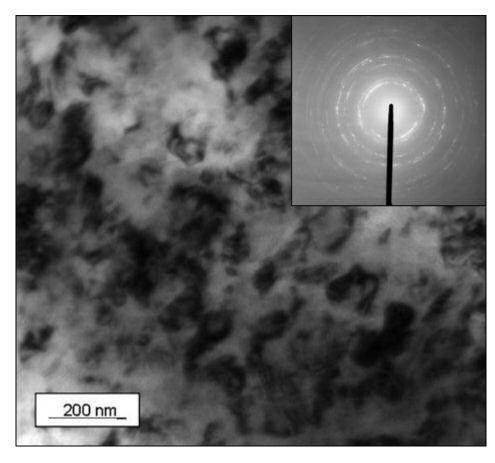
- Figure 1. Comparison of the peak widths of AlMg4.8 in the as-milled (8 h) and annealed (400 °C / 3 h) state obtained by XRD
- Figure 2. TEM bright field image of the as-milled nc AlMg4.8 powder and selected area diffraction pattern
- Figure 3. Change of micro-hardness and crystallite size as a function of milling time
- Figure 4. Crystallite size and dislocation density evolution in dependence of annealing for 3 h at different
- temperatures of AlMg4.8 preliminary milled for 8 h
- Figure 5. Hardness of AlMg4.8 powder milled for 8 h after annealing at different temperatures for 0.5 h, 3 h and 100 h
- Figure 6. DSC curves for the as-milled and annealed samples at a heating rate of 10 K/min
- Figure 7. Correlation of grain size and hardness of AlMg4.8 linearised in a Hall-Petch plot after different process routes
- Figure 8. Hall Petch constant k depends on the Mg content; k value of 4.8 wt% was calculated in the present work
- while the other results are reported in literature [37-39].
- Figure 9. Kissinger plot indicating the activation energy for grain growth
- Figure 10. Arrhenius plot of rate constant values obtained by the growth results

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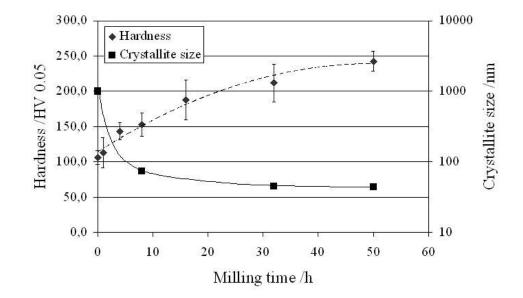
Hardening and thermal stability of NC AlMg4.8



200x154mm (150 x 150 DPI)

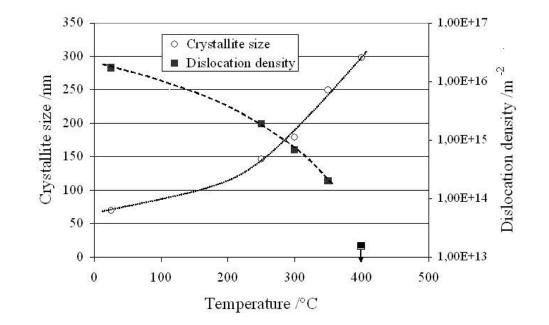


TEM bright field image of the as-milled nc AlMg4.8 powder and selected area diffraction pattern 165x152mm (72 x 72 DPI)



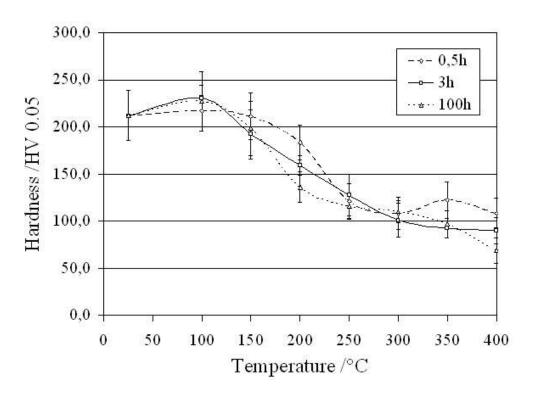
Change of micro-hardness and crystallite size as a function of milling time. 184x114mm (100 × 100 DPI)

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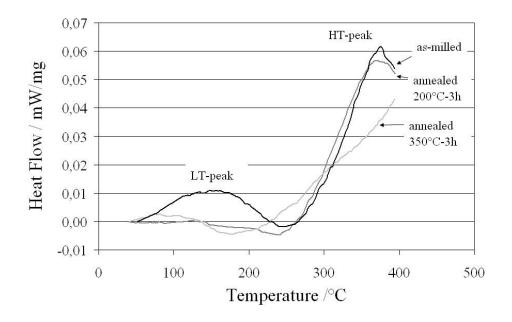
Crystallite size and dislocation density evolution in dependence of annealing for 3 h at different temperatures of AlMg4.8 preliminary milled for 8 h. 180x114mm (100 x 100 DPI)



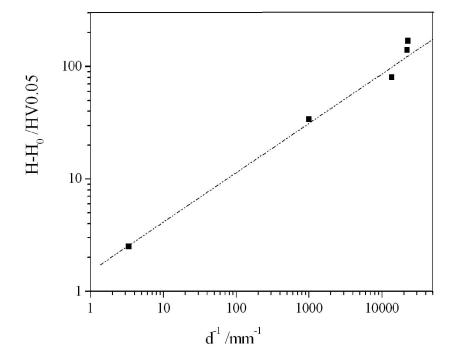


Microhardness of AlMg4.8 powder milled for 8 h after annealing at different temperatures for 0.5 h, 3 h and 100 h.

149x109mm (100 x 100 DPI)



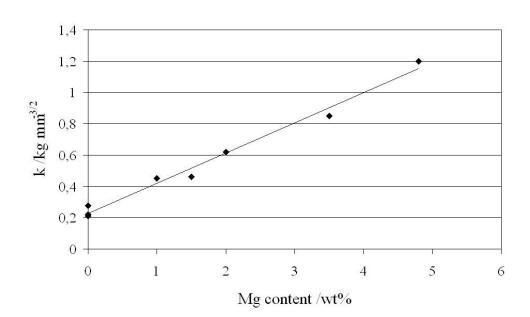
DSC curves for the as-milled and annealed samples at a heating rate of 10 K/min



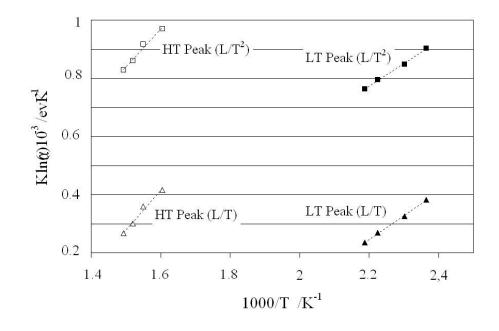
Correlation of grain size and hardness of AlMg4.8 linearised in a Hall-Petch plot after different process routes 279x215mm (150 × 150 DBI)

279x215mm (150 x 150 DPI)

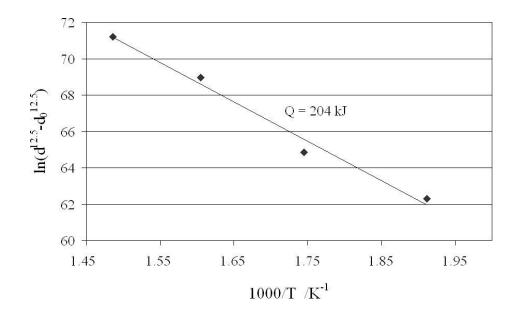




Hall Petch constant k depends on the Mg content; k value of 4.8 wt% was calculated in the present work while the other results are reported in literature [37-39] 342x213mm (72 x 72 DPI)



Kissinger plot indicating the activation energy for grain growth



Arrhenius plot of rate constant values obtained by the growth results 342x213mm (72 x 72 DPI)

Elements	Si	Fe	Cu	Mn	Mg	Cr	Ni	Ti	Al
AlMg4.8-0/wt%	0.045	0.070	0.007	0.002	4.77	0.002	0.01	0.005	Balance
as-sprayed									
AlMg4.8-32 /wt%	0.048	0.078	0.005	0.003	4.76	0.002	0.01	0.005	Balance
after 32 milling									

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Milling time /h	H /ppm	N /ppm	O /ppm
0/8/32	36.5 / 44 / 62	11 / 22 / 45	581 / 1117 / 2009

Heating rate /K/min	5	10	20	30
T P1	149,6	161,1	176,3	183,9
T P2	349,6	371,8	384,4	396,3

Peak designation	LT-peak		HT-peak	
Approach	α=L/T	$\alpha = L/T^2$	α=L/T	$\alpha = L/T^2$
eV	1,30	1,24	2,16	2,08
kJ/mol	126,03	120,20	209,47	200,92