

A continuous mapping of sleep states through association of EEG with a mesoscale cortical model

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Abstract Here we show that a mathematical model of the human sleep cycle can be used to obtain a detailed description of electroencephalogram (EEG) sleep stages, and we discuss how this analysis may aid in the prediction and prevention of seizures during sleep. The association between EEG data and the cortical model is found via locally linear embedding (LLE), a method of dimensionality reduction. We first show that LLE can distinguish between traditional sleep stages when applied to EEG data. It reliably separates REM and non-REM sleep and maps the EEG data to a low-dimensional output space where the sleep state changes smoothly over time. We also incorporate the concept of strongly connected components and use this as a method of automatic outlier rejection for EEG data. Then, by using LLE on a hybrid data set containing both sleep EEG and signals generated from the mesoscale cortical model, we quantify the relationship

between the data and the mathematical model. This enables us to take any sample of sleep EEG data and associate it with a position among the continuous range of sleep states provided by the model; we can thus infer a trajectory of states as the subject sleeps. Lastly, we show that this method gives consistent results for various subjects over a full night of sleep and can be done in real time.

Keywords Sleep · Sleep staging · Sleep scoring · Seizure · Locally linear embedding · Electroencephalogram · Cortical model · Mesoscale · Mean-field

1 Introduction

The standard method of sleep scoring involves categorization of electroencephalogram (EEG) data into five separate stages (Niedermeyer and da Silva 2005). However, the *discrete* nature of these stages limits their utility as analytical and predictive tools. For example, in a study of human epilepsy, it may be observed that a seizure occurred during stage 2 sleep. This prompts further questions: Was the subject descending to deeper stages of sleep or arising from them? How quickly was the subject moving through each stage? Was a transition imminent when the seizure occurred?

The use of a mathematical model of the human sleep cycle may allow us to answer such questions by providing a *continuous* spectrum of sleep states, ranging from REM to the deepest slow-wave sleep. If the model can be directly associated with human sleep EEG data, it will be possible to track the subject's state to identify

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the stage as well as changes in sleep depth and proximity to transitions. Ideally, this would be done in real-time, where the state is continuously determined as the subject sleeps. The process must be consistent over various subjects and robust to non-standard sleep cycles and periods of waking.

Here we utilize a technique called locally linear embedding (LLE) to make this connection between a model of the human sleep cycle and EEG data. First, we present a model of the human cortex with subcortical inputs represented by added driven noise, and we describe the associated mathematical representation of the sleep cycle (Section 2). We then introduce the technique of locally linear embedding (Section 3) and show that it provides the ability to distinguish between sleep stages when applied to EEG data (Section 4). These results demonstrate reliable separation between REM and NREM sleep data and provide a smooth temporal progression through the various stages of sleep. We also present the concept of strongly connected components as a method of outlier rejection for EEG data (Section 3.2) and introduce a method for automatic selection of LLE parameters (Section 4.3). Then, by performing LLE on a hybrid data set containing both sleep EEG and signals generated from the mathematical model, we are able to integrate the EEG and the model (Section 5). This allows us to take any sample of sleep EEG data and determine its position within the continuous range of sleep states provided by the model. We show that this method provides consistent results for various subjects over a full night of sleep, and it could be done online as the subject sleeps.

2 Mean-field cortical model

2.1 Background and mathematics

Mean-field models of the cortex are well-suited to the study of brain states described by EEG signals, including sleep. The variables in these models, representing quantities that are averaged over the millimeter scale, are comparable to the mesoscale measurements of EEG electrodes. More specifically, we choose a cortical model developed most recently in Liley et al. (2002) and Steyn-Ross et al. (1999, 2003). In addition to sleep, it has been used to model epileptic seizures (Kramer et al. 2005), anesthesia (Steyn-Ross et al. 2004; Bojak and Liley 2005), and the transition to seizure due to application of anesthetic agents (Liley and Bojak 2005).

Here, we use the dimensionless formulation of the model as described in Kramer et al. (2007), with two parameters Δh_e^{rest} and L added to represent neuromod-

ulators that regulate the natural sleep cycle, as was done in Steyn-Ross et al. (2005):

$$\frac{\partial \tilde{h}_e}{\partial \tilde{t}} = 1 - \tilde{h}_e + \frac{\Delta h_e^{rest}}{h^{rest}} + L \Gamma_e \left(h_e^0 - \tilde{h}_e \right) \tilde{I}_{ee} + \Gamma_i \left(h_i^0 - \tilde{h}_e \right) \tilde{I}_{ie}, \quad (1)$$

$$\frac{\partial \tilde{h}_i}{\partial \tilde{t}} = 1 - \tilde{h}_i + L \Gamma_e \left(h_e^0 - \tilde{h}_i \right) \tilde{I}_{ei} + \Gamma_i \left(h_i^0 - \tilde{h}_i \right) \tilde{I}_{ii}, \quad (2)$$

$$\left(\frac{1}{T_e} \frac{\partial}{\partial \tilde{t}} + 1 \right)^2 \tilde{I}_{ee} = N_e^\beta \tilde{S}_e \left[\tilde{h}_e \right] + \tilde{\phi}_e + P_{ee} + \tilde{\Gamma}_1, \quad (3)$$

$$\left(\frac{1}{T_e} \frac{\partial}{\partial \tilde{t}} + 1 \right)^2 \tilde{I}_{ei} = N_e^\beta \tilde{S}_e \left[\tilde{h}_e \right] + \tilde{\phi}_i + P_{ei} + \tilde{\Gamma}_2, \quad (4)$$

$$\left(\frac{1}{T_i} \frac{\partial}{\partial \tilde{t}} + 1 \right)^2 \tilde{I}_{ie} = N_i^\beta \tilde{S}_i \left[\tilde{h}_i \right] + P_{ie} + \tilde{\Gamma}_3, \quad (5)$$

$$\left(\frac{1}{T_i} \frac{\partial}{\partial \tilde{t}} + 1 \right)^2 \tilde{I}_{ii} = N_i^\beta \tilde{S}_i \left[\tilde{h}_i \right] + P_{ii} + \tilde{\Gamma}_4, \quad (6)$$

$$\left(\frac{1}{\lambda_e} \frac{\partial}{\partial \tilde{t}} + 1 \right)^2 \tilde{\phi}_e = \frac{1}{\lambda_e^2} \frac{\partial^2 \tilde{\phi}_e}{\partial \tilde{x}^2} + \left(\frac{1}{\lambda_e} \frac{\partial}{\partial \tilde{t}} + 1 \right) N_e^\alpha \tilde{S}_e \left[\tilde{h}_e \right], \quad (7)$$

$$\left(\frac{1}{\lambda_i} \frac{\partial}{\partial \tilde{t}} + 1 \right)^2 \tilde{\phi}_i = \frac{1}{\lambda_i^2} \frac{\partial^2 \tilde{\phi}_i}{\partial \tilde{x}^2} + \left(\frac{1}{\lambda_i} \frac{\partial}{\partial \tilde{t}} + 1 \right) N_i^\alpha \tilde{S}_i \left[\tilde{h}_i \right]. \quad (8)$$

The model contains two groups of equations: one that describes the evolution of the excitatory population (Eqs. (1), (3), (5), (7)) and one that governs the inhibitory population (Eqs. (2), (4), (6), (8)). Each variable is a function of dimensionless space (\tilde{x}) and time (\tilde{t}), and the subscript denotes its association with the excitatory or inhibitory population. For example, in the excitatory population, the mean soma potential is represented by \tilde{h}_e , while \tilde{I}_{ie} is the input current from population i to population e . The synaptic currents are functions of local input, e.g. $N_e^\beta \tilde{S}_e$ where

$$\tilde{S}_e \left[\tilde{h}_e \right] = \frac{1}{1 + \exp \left[-\tilde{g}_e \left(\tilde{h}_e - \tilde{\theta}_e \right) \right]}; \quad (9)$$

this function converts the potential of the excitatory population into a mean firing rate. Synaptic currents are also affected by long-range corticocortical input $\tilde{\phi}_e$ and subcortical stochastic inputs such as $\tilde{\Gamma}_1$, which we define to be a function of zero-mean Gaussian white noise ξ_1 :

$$\tilde{\Gamma}_1 = \alpha_{ee} \sqrt{P_{ee}} \xi_1 \left[\tilde{x}, \tilde{t} \right]. \quad (10)$$

Here α_{ee} is a constant that determines the variance of the stochastic input. Please refer to Table 1 for further descriptions of all variables and parameters.

For completeness, we will include the full model in our simulations; however, it should be noted that a reduced version would suffice in this case. For example, we will utilize only the temporal evolution of variables, so it would be possible to convert the model to a system of ODEs by removing the spatial derivatives from Eqs. (7) and (8). In addition, the subdivision of local excitatory inputs, represented by $N_e^\beta \tilde{S}_e[\tilde{h}_e]$ in Eqs. (3) and (4), is unnecessary. Making these changes would perhaps reduce the computation time for numerical solutions to the model, but we would not expect them to affect the results.

For the purpose of modeling sleep, we will focus on the parameters L and Δh_e^{rest} and the variable \tilde{h}_e . The parameters represent the actions of neuromodulators adenosine and acetylcholine (ACh) that aid in the regulation of the human sleep cycle. Adenosine reflects the activity of the homeostatic drive to sleep, which is modulated by various somnogens. The ACh input into the cortex is a measure of the activity of the various brain stem controllers of sleep. Note that we have not specifically modeled the complex intrinsic interactions between the various brain stem nuclei. In this paper, we are primarily concerned with the interaction of their neuromodulator output with the cerebral cortex and thus model their effects only as extrinsic alterations in ACh.

In general, adenosine acts to reduce the resting potential of excitatory cells, thus making them less likely to fire; ACh does the opposite by raising the resting potential. These changes are represented in the model by Δh_e^{rest} , which adds directly to the resting potential of the excitatory population (disguised as a “1” in the dimensionless equations). In addition, ACh decreases the amplitude of the excitatory postsynaptic potential, effectively reducing the synaptic gain. In the model, this corresponds to a reduction in the effect of synaptic currents \tilde{I}_{ee} and \tilde{I}_{ei} ; therefore, the parameter L is multiplied by these quantities to simulate a change in synaptic gain. Lastly, as was done in Steyn-Ross et al. (2005), we take the mean excitatory soma potential \tilde{h}_e to be representative of cortical activity; we will compare this variable to EEG measurements using locally linear embedding.

2.2 Model of the human sleep cycle

The mechanisms underlying human sleep and waking are complex; for recent, detailed reviews of the brain stem and hypothalamic control of sleep in thalamo-cortical systems see Fuller et al. (2006, 2007), McCarley (2007), Rosenwasser (2009), Saper et al. (2005a, b). In summary, the wakeful state may be characterized by high levels of activity in aminergic, cholinergic, orexinergic and glutamatergic neuronal populations in the brain stem and hypothalamus. The overall effect is to maintain the thalamo-cortical neurons in a depolarized,

Table 1 Dimensionless variables and parameters of the SPDE cortical model

Symbol	Description	Typical value
$\tilde{h}_{e,i}$	Spatially averaged soma potential for neuron populations	–
$\tilde{I}_{ee,ei}$	Postsynaptic activation due to excitatory inputs	–
$\tilde{I}_{ie,ii}$	Postsynaptic activation due to inhibitory inputs	–
$\tilde{\phi}_{e,i}$	Long-range (corticocortical) input to e and i populations	–
\tilde{t}	Time (dimensionless)	–
\tilde{x}	Space (dimensionless)	–
$\Gamma_{e,i}$	Influence of synaptic input on mean soma potential	4.6875×10^{-4} , 0.0105
$h_{e,i}^0$	Reversal potential	0, 1.0938
$T_{e,i}$	Neurotransmitter rate constant	12.0, 3.6
$\lambda_{e,i}$	Inverse length scale for corticocortical connections	11.2, 11.2
$P_{ee,ei}$	Subcortical input from excitatory population	25.0, 25.0
$P_{ie,ii}$	Subcortical input from inhibitory population	25.0, 25.0
$N_{e,i}^\alpha$	Number of distant (corticocortical) connections from excitatory populations to e and i populations	3710, 3710
$N_{e,i}^\beta$	Number of local synaptic connections from e and i populations	410, 800
$\tilde{g}_{e,i}$	Slope at inflection point of sigmoid function \tilde{S}_e	–29.021, –19.347
$\tilde{\theta}_{e,i}$	Inflection point for sigmoid function \tilde{S}_e	0.91406, 0.91406

Values for the dimensional parameters were taken from Wilson et al. (2006), with the exception of γ_i which was chosen to be $90s^{-1}$. The dimensionless parameters were then calculated as described in Kramer et al. (2007)

active, and continually firing state. These excitatory neurons also inhibit activity in various gamma-aminobutyric-acid (GABA)ergic cell populations, particularly in the ventro-lateral pre-optic area (VLPO), basal forebrain, and in the reticular nucleus of the thalamus. With the build up of homeostatic and circadian pressure to sleep (possibly mediated by various activity-dependent somnogens such as adenosine), the wake-active neurons are inhibited, which then allows the sleep-promoting neurons of the VLPO to start firing and trigger the transition from wakefulness to NREM sleep. This results in quiescence of the aminergic, orexinergic, and cholinergic brain-stem neuromodulator centers; which in turn allows hyperpolarization of the cortico-thalamic systems and hence the burst firing patterns characteristic of slow wave sleep. If these neurons are only moderately hyperpolarized, the EEG is dominated by the sleep spindles and K-complexes characteristic of stage 2 sleep. With more profound hyperpolarization the EEG is dominated by the delta waves of stages 3 and 4 (Steriade and Amzica 1998; Steriade and Timofeev 2001). This progressive slowing of the dominant frequency is captured by measures such as the permutation entropy index (Olofsen et al. 2008). The transition from NREM to REM sleep is associated with cortico-thalamic depolarization caused by activation of cholinergic and glutamatergic brain stem systems (mainly in and near the pedunculo-pontine tegmentum). The neuromodulatory environment of REM sleep differs from the wakeful state in that the amines and orexinergic systems are inactive in REM sleep, but active in the wakefulness; however, this distinction is not explicit in the present model.

Mathematically, we follow Steyn-Ross et al. (2005), where the representation of the sleep cycle is based on changes in neuromodulators L and Δh_e^{rest} . In order to visualize this, we look at steady-state solutions of h_e (without stochastic input) as L and Δh_e^{rest} are varied; these solutions create what we will refer to as the “sleep manifold” (Fig. 1). Notice that, for most parameter values, there is only one steady state solution. However, in certain cases, there are three solutions (two stable and one unstable), causing the manifold to fold over on itself. This fold is seen on the left side of Fig. 1. In this model, the top branch of solutions on the manifold is intended to be representative of REM sleep. Starting at this point, we can imagine that during sleep, there is a gradual descent to deep slow-wave sleep by following a trajectory down the right side of the manifold where there is only one steady state solution. This happens in a smooth continuous manner. Then the quick transition from slow-wave sleep to REM is simulated by a jump across the fold from the bottom branch

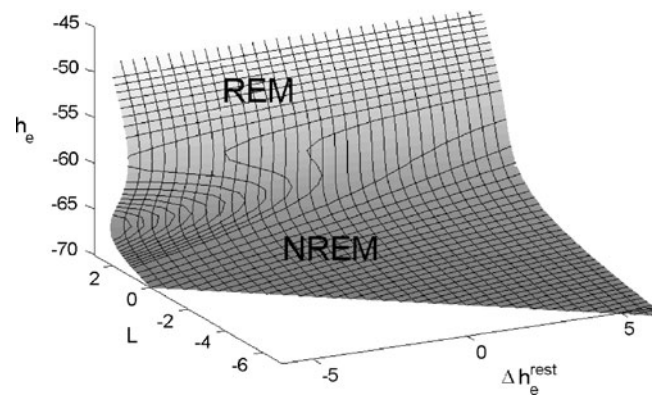


Fig. 1 The manifold of steady states in h_e from the mesoscale cortical model, hereafter referred to as the “sleep manifold.” The parameters L and Δh_e^{rest} represent the actions of adenosine and acetylcholine, neuromodulators that vary over the course of the human sleep cycle. The manifold has two stable solutions on its left side; a jump from the bottom solution to the top solution represents the fast transition between NREM and REM sleep. However, the slow transition from REM to NREM occurs smoothly down the right side of the manifold, where there is only one set of solutions

of solutions to the top branch. This mimics the rapid transition from deep sleep to REM that is observed in human EEG recordings. This process of gradually moving from REM to NREM sleep and then quickly jumping back to REM represents one sleep cycle in the model.

This model has been previously studied. Steyn-Ross et al. (2005) calculated the EEG total power, fractions of high and low power, and correlation time exhibited by the model at the transition from slow-wave sleep to REM; it was found that they qualitatively matched both human clinical sleep recordings and cortical measurements from a cat. The model was also studied in two spatial dimensions to investigate stable oscillatory states similar to slow-wave sleep, and it was shown that a transition from one state to another can occur due to stochastic fluctuations (Wilson et al. 2005). Lastly, Wilson et al. (2006) interpreted the k-complex as a transient shift from a stable low-firing state to an unstable high-firing state and used this model to demonstrate the mechanism by which the transition may occur.

Because we are interested in comparing this model directly to human EEG recordings, we will use the sleep manifold as a way to generate model “EEG-like” signals. We will choose values of L and Δh_e^{rest} , find the numerical solution of the model for a given length of time, convert the dimensionless \tilde{h}_e to mV, and downsample it to match the EEG recordings. By doing this for many different values of L and Δh_e^{rest} we can obtain representative signals of every sleep stage.

It has previously been argued that \tilde{h}_e cannot be directly compared to measurements from cortical surface or scalp electrodes because those measurements are based on extracellular current flow, as opposed to the soma potential. This is important for the modeling of certain cortical phenomena; for example, in performing simulations of feedback control for the suppression of epileptic seizures, the value of the electrode measurement is fed directly back to the cortex to affect \tilde{h}_e , with little or no time delay (Lopour and Szeri 2010). In that case, the relationship between \tilde{h}_e and the electrode measurement at any given time is very important. However, in the present analysis of EEG data using LLE, we are only interested in matching *scaled features* of the data that are calculated over 30-second intervals. We will not attempt to compare the temporal progression of \tilde{h}_e directly to the EEG data. The previous work mentioned above has demonstrated a correspondence between \tilde{h}_e and sleep EEG data with regard to these general features, so we feel confident in using it for our analysis without the addition of a scalp electrode model.

3 Locally linear embedding (LLE)

Locally linear embedding is a method of nonlinear dimensionality reduction that was originally introduced in Roweis and Saul (2000). It is useful for visualizing high-dimensional data sets as they would be embedded in a low-dimensional space, and it can often uncover relationships and patterns that are masked by the complexity of the original data set. It has been used to obtain maps of facial expressions and classify handwritten digits (Saul et al. 2003), as well as discriminate between normal and pre-seizure EEG measurements (Ataee et al. 2007). Here we will use LLE to characterize sleep EEG data and the numerical solutions of the cortical model. By embedding both in a two-dimensional space, we will be able to associate traditional EEG sleep stages with the continuous spectrum of states provided by the model.

3.1 The algorithm

Let us begin with a high-dimensional data set stored in a matrix \mathbf{X} of size $D \times N$, where each column \mathbf{X}_i represents one of the N D -dimensional data points. Then the LLE algorithm consists of three steps:

1. *Calculate the nearest neighbors of each data point \mathbf{X}_i in the D -dimensional space.* This can be done in several ways; for example, we might choose the k

closest points based on Euclidian distance, or we may choose only the points within a sphere of a given radius.

2. *Determine the best reconstruction of each point using only its nearest neighbors.* Mathematically, this takes the form of a least squares minimization problem:

$$\min_W \sum_{i=1}^N \left\| \mathbf{X}_i - \sum_{j=1}^k W_{ij} \mathbf{X}_j \right\|^2, \quad (11)$$

where k represents the number of nearest neighbors. Our goal is to choose the weights W that best reconstruct the original data points in the D -dimensional space, based on the criteria of least-squared error. Because we use only the nearest neighbors, we must have $W_{ij} = 0$ if \mathbf{X}_j is not a neighbor of \mathbf{X}_i . In addition, we guarantee invariance to translations by enforcing $\sum_j W_{ij} = 1$. Note that the minimization can be calculated individually for every i .

3. *Compute the low-dimensional output vectors \mathbf{Y}_i .* These are chosen to provide the best global reconstruction using the weights W from the previous step. Again, this can be formulated as a least squares minimization:

$$\min_{\mathbf{Y}} \sum_{i=1}^N \left\| \mathbf{Y}_i - \sum_{j=1}^k W_{ij} \mathbf{Y}_j \right\|^2. \quad (12)$$

Here we are making the assumption that the weights that give the best reconstruction in D dimensions will also be the optimal weights in the lower-dimensional space. In this case, the N minimization problems are coupled by the elements of \mathbf{Y} , so they must be solved simultaneously.

A detailed description of the algorithm and several examples are provided in Saul et al. (2003). In addition, a Matlab implementation of LLE is available on the authors' website (Roweis and Saul 2009); it was used to generate all results presented here.

As a simple example, consider using LLE on a known 3D manifold (Fig. 2). In this toy example, the underlying manifold is known (although normally this would not be the case), and we recognize that it has only two dimensions, despite living in 3-dimensional space as shown in Fig. 2(a). The data set \mathbf{X} consists of a random sampling of points from the manifold (Fig. 2(b)), and the LLE output for a reduction to two dimensions is displayed in Fig. 2(c). Here we see that LLE successfully unravels the manifold and uncovers its true 2D nature.

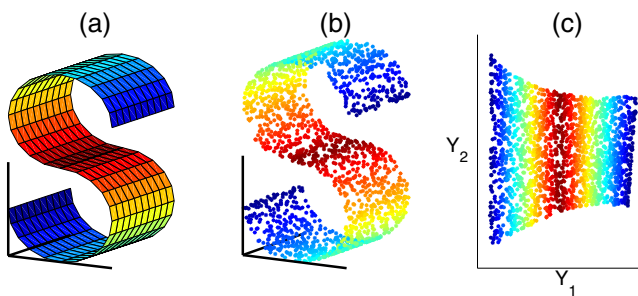


Fig. 2 A simple example of LLE, where three dimensions are reduced to two. **(a)** The underlying manifold, which lives in 3D space but has only two dimensions. In a typical LLE problem, the shape of this manifold is unknown and has too many dimensions to visualize easily. **(b)** A sampling of points from the manifold, which serves as the input to the LLE algorithm. **(c)** The result of applying LLE to the data set in **(b)**. Note that in the $Y_1 - Y_2$ output space, the manifold has been flattened to reveal its two principal dimensions. This figure was generated using the “scurve.m” code from the LLE website (Roweis and Saul 2009)

A possible source of confusion with locally linear embedding is the interpretation of output dimensions such as Y_1 and Y_2 . Unlike linear methods such as principal component analysis, LLE does not provide a description of the output vectors in terms of the original D dimensions. The elements of \mathbf{Y} are chosen to give the best *local* reconstructions based on a global minimization problem; this means that the interpretation of \mathbf{Y} is different for every data point, and it cannot be described by a simple combination of the original dimensions.

3.2 Strongly connected components

The use of the LLE algorithm is based on the assumption that the entire data set lies on the same manifold in high-dimensional space. If more than one manifold is present, the locally linear reconstructions will no longer be accurate (imagine, for example, a point with nearest neighbors located on two separate manifolds). Therefore, before using LLE on a data set, we must verify this assumption.

The mathematics and terminology of directed graphs allows us to accomplish this task (Tarjan 1972). Note that when we calculate the nearest neighbors in the first step of the LLE algorithm, we create a directed graph based on the data points. For example, suppose there is a data set of seven points, and we have determined that point 2 is a neighbor of point 1, point 5 is a neighbor of point 2, etc. This can be depicted by arrows drawn from each point to its neighbors (Fig. 3). Then we can define a *strongly connected component* as a group of points where the arrows created by nearest neighbor

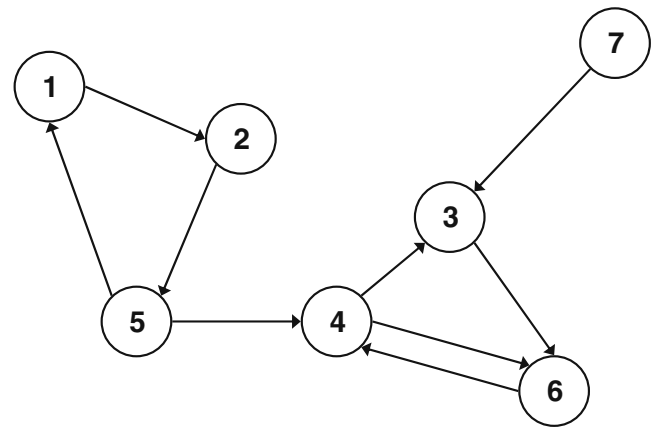


Fig. 3 An example of a directed graph generated by nearest neighbor associations. Here point 2 is a neighbor of point 1, point 3 is a neighbor of point 7, etc. In this case, the directed graph forms two strongly connected components: points 1, 2, 5, and points 3, 4, 6. In analyzing this data set, we would use LLE separately on each of these components and would remove point 7, which is not strongly connected to any other point

associations allow for travel from every point in the group to every other point in the group (Tarjan 1972). *When a group of data points is strongly connected, this indicates that they lie on the same manifold* (Saul et al. 2003).

The example in Fig. 3 has two strongly connected components: points 1, 2, 5, and points 3, 4, 6. However, the two groups are not strongly connected together; one can move from the first group to the second through the connection between 4 and 5, but there is no way to get from the second group to the first. Point 7 is not strongly connected to any other point. Therefore, to use LLE on this sample data set, we would remove point 7 and use the algorithm separately on each strongly connected component.

There are several ways to identify the strongly connected components of a data set. The most traditional method involves an algorithm based on depth-first search of the directed graph (Tarjan 1972). An alternative method relies on analysis of the eigenspace that results from the LLE calculations (Polito and Perona 2001). It is also true that choosing the nearest neighbors in a different manner or increasing the value of k can change the structure of the strongly connected components. However, for the purposes of this study, we used the MATLAB function `dmperm` on a matrix containing the nearest neighbor associations for the data set. This function, based on the Dulmage-Mendelsohn decomposition, permutes the rows and columns of a matrix to put it into block diagonal form; by including the fact that every point is a neighbor with itself, we can guarantee that this permutation will be symmetric.

As output, `dmperm` provides the new order of rows and columns and identifies the blocks of the permuted matrix, where each block represents one strongly connected component within the data.

A remark about principal component analysis (PCA) is in order. This is perhaps the most common mode of dimensionality reduction, and it has also been used in the analysis of sleep EEG data (Gervasoni et al. 2004; Jobert et al. 1994; Corsi-Cabrera et al. 2000). However, PCA places the greatest importance on the directions of largest variance and relies on the assumption that the data is best reconstructed by a *linear* combination of the original measurements. While we tried PCA and achieved reasonable results, the nonlinear nature of the sleep manifold suggests that a more sophisticated solution is necessary. In addition, the concept of nearest neighbors on which the LLE algorithm is based enabled improvement in the separation of different sleep stages (see Section 4.4), and it played a crucial role in defining the quantitative relationship between the EEG data and mathematical model, as is discussed in Section 5.2.

4 LLE applied to sleep EEG data

Before examining the connection between EEG data and the mathematical model of the sleep cycle, we will first discuss the results of applying LLE to sleep EEG only. After introducing the data sets and our methods, we show that LLE can separate EEG data by sleep stage and provide a continuous representation of sleep depth.

4.1 Sleep EEG data

The EEG data used for this analysis was obtained from the Sleep-EDF database (Kemp 2009), which is part of the PhysioBank online resource of physiologic signals for biomedical research (Goldberger et al. 2000). We used four data sets (sc4002e0, sc4012e0, sc4102e0, and sc4112e0), each one consisting of a European data format (EDF) file and a file containing the hypnogram data. They were converted to ASCII format and then imported into Matlab.

The data were gathered in 1989 from healthy males and females between the ages of 21 and 35. Recordings were obtained over the course of one full day and include horizontal electrooculogram (EOG), two channels of EEG (Fpz-Cz and Pz-Oz sampled at 100 Hz), submental-electromyogram (EMG) envelope, oronasal airflow, and rectal body temperature. However, we used only the data from the Fpz-Cz EEG electrode

pair in our analysis. The hypnogram data was generated via manual scoring according to Rechtschaffen & Kales using the two channels of EEG. For more details on the subjects, recording methods, and sleep staging, please see the full description in Mourtazaev et al. (1995).

4.2 LLE input based on EEG features

In order to use the EEG as an input to the LLE function, we need to define our high-dimensional data set. We do this by dividing the signal into non-overlapping windows and calculating both statistical and frequency-based *features* for each one. Therefore each window becomes one high-dimensional data point, where the dimension equals the number of features. Because the data was scored using 30-second epochs, this was a natural choice for the window length. Thus, if we have 100 minutes of EEG data and we calculate six features, we will input 200 six-dimensional points into LLE and seek the embedding in two dimensions.

We start with a pool of 17 features and use various subsets to perform the LLE analysis. An algorithm for the automated choice of feature combinations is discussed in Section 4.3. The 17 features are as follows:

Power in different frequency bands This group of five features consists of total power in the delta (up to 4 Hz), theta (4–7.5 Hz), alpha (7.5–12 Hz), beta (12–26 Hz), and gamma ranges (above 26 Hz).

Total power This is the total power in all five frequency bands.

Statistical measures These include variance, skewness, and kurtosis. Whereas the variance captures the spread of the data and is always positive, skewness is a measure of the asymmetry around the sample mean, i.e. negative skewness indicates that more data points lie below the mean than above. Kurtosis is a measure of how prone the distribution is to outliers; a signal with high kurtosis has infrequent large deviations from the mean.

Spindle score The spindle detector identifies segments of the EEG signal where the difference between consecutive points changes from positive to negative five times in a row, thereby creating two peaks and two troughs. The lag parameter τ_L defines the number of sample points spanned by each rise or fall within the sought-for spindle, so it can be adjusted to search for these motifs at lower frequencies. We set $\tau_L = 5$, which allows for detection of 8–12 Hz spindles in data sampled at 100 Hz (with the maximum response occurring for spindles at 10 Hz), and we used a minimum threshold of zero. The overall spindle score indicates the percentage of the

signal that was classified as spindle activity. Matlab code for this function is provided in McKay et al. (2010).

Permutation entropy Similar to the spindle score, the permutation entropy (PE) identifies motifs in the EEG data, such as peaks, troughs, and slopes. The PE has its maximum value when there is an equal distribution of all motifs and its minimum value when only a single motif is present. In this way, it is a measure of the “flatness” or “uncertainty” of the signal. Here we use the composite permutation entropy index (CPEI), which combines the PE with $\tau_L = 1$ and $\tau_L = 2$ with a minimum threshold level. In our study, we set the threshold at 1% of the interquartile range of the EEG data. Further descriptions of this measure and an associated MATLAB function can be found in Olofsen et al. (2008).

The CPEI has been found to be a good measure of anesthetic depth, and the motif-based methods used for permutation entropy and spindle detection are generally robust to noise. This is demonstrated in Olofsen et al. (2008), where the CPEI is calculated for both a time-varying signal and the same signal with added white noise of various magnitudes. As mentioned above, the noise threshold for PE is built into the calculation. These reasons (and the availability of published MATLAB code) led us to choose motif-based methods over more common parametric measures.

Properties of log power These four features are based on the log of the power spectral density (PSD), as obtained by Welch’s method. First, we omit the delta and alpha peaks and calculate the slope and offset of a linear fit. We then determine the maximum value of the PSD above the linear estimate in the alpha range (8–17 Hz) and the maximum value of the PSD in the delta range (0.5–4 Hz). These values will generally be large when a prominent peak is present. The code for generating these features was based on a Matlab function found in Leslie et al. (2009).

Power fractions The low power fraction is obtained by summing the power in the delta and theta ranges and dividing by total power. Similarly, the high power fraction is calculated by summation of the power in the beta and gamma ranges and dividing by total power.

After the initial calculation, each feature was divided by its root mean square (RMS) value.

The selection of a subset of features from this list may seem like a difficult task. It is certainly an important one—the use of all 17 features or a “nonsensical”

subset will give poor results. However, it is worth noting that there are *many* combinations that result in a satisfactory separation between sleep stages in the LLE embedding. While each one may be slightly different, there will be a large number of high quality with respect to discrimination.

4.3 Automated ranking of feature sets

When we apply LLE to the EEG data, there are essentially only three choices that we must make:

1. *How many nearest neighbors should we include?* In other words, what is the value of k ? The LLE embedding will be stable over a range of values; we generally expect that k will be greater than the number of output dimensions and smaller than the original number of dimensions D (Saul et al. 2003).
2. *What should be the dimensionality of the LLE output space?* A nice property of the LLE algorithm is that each dimension is preserved as additional dimensions are added. Therefore, if we look at the results in two dimensions and do not achieve the desired mapping, we can add a third dimension without affecting the first two.
3. *Which combination of features should we use?* Employing all 17 features in our LLE analysis does not guarantee good separation between sleep stages because some of the features may not show consistent variation as the sleep depth changes. In addition, some features, such as the variance and the power in the delta band, show similar trends; we may achieve better results by eliminating these redundancies.

In this section, we focus on the last of these questions.

While we were able to identify many effective feature combinations through educated guesswork, we wanted to evaluate the utility of LLE as a method of sleep staging by identifying the best possible results. In this case, the “best” results are those that provide a large separation between sleep stages, especially between REM and deep slow-wave sleep. Because testing each combination of the features is an onerous task, e.g. choosing six features from a pool of 17 results in 12376 combinations, we developed an algorithm to evaluate the results automatically. It first identifies two groups of points: those marked as REM in the hypnogram and those determined to be stage 4. It then tracks two parameters based on the separation between those two groups of data points as they are embedded in the LLE output space.

We first measure the percent separation between REM and stage 4, calculated as

$$a_i = 100 \cdot \operatorname{erf} \left(\sqrt{0.5} \frac{\mu_4 - \mu_{REM}}{\sigma_4 + \sigma_{REM}} \right), \quad (13)$$

where μ and σ are the mean and standard deviation, respectively. This is based on the assumption that the best separation occurs when the distance between the means is large and the total standard deviation is small. We perform this calculation in both the Y_1 and Y_2 directions and combine those measurements using the 2-norm to obtain the first parameter:

$$A = \sqrt{a_{Y_1}^2 + a_{Y_2}^2}. \quad (14)$$

The second parameter B uses the concept of nearest neighbors to evaluate separation; for example, if the stage 4 data points have only other stage 4 points as nearest neighbors, then we can infer that they are completely separated from the other sleep stages. More specifically, it measures the number of stage 4 points with REM points as nearest neighbors and divides that by the total number of stage 4 points. If the stage 4 group is isolated, we will have $B = 0$.

We determined the values of A and B for all possible combinations of six features. There were 267 feature sets where A exceeded a threshold of 90% separation in each direction: $A > \sqrt{90^2 + 90^2}$. We then identified the 267 feature sets with the lowest values of B . By finding the combinations that were common to both groups, we identified the 11 best feature sets. Visual inspection of the LLE results for these combinations confirmed the desired separation between REM and stage 4 sleep. Note that all 11 of these feature combinations provided results with $B = 0$.

4.4 Separation of sleep stages via LLE

Having described the EEG data set using frequency-based and statistical measures and having identified the most effective subsets of those features, we are now ready to apply the LLE algorithm. As a representative result, we choose one of the 11 feature sets from the previous section; the six features are power in the delta and theta bands, variance, spindle score, maximum height of the PSD above a linear estimate in the alpha band, and high power fraction. These are plotted in Fig. 4 for 178 epochs from the sc4002e0 data set. The corresponding hypnogram is included for reference. Note that the features were calculated in 30-second non-overlapping windows to match the sleep scoring of the hypnogram.

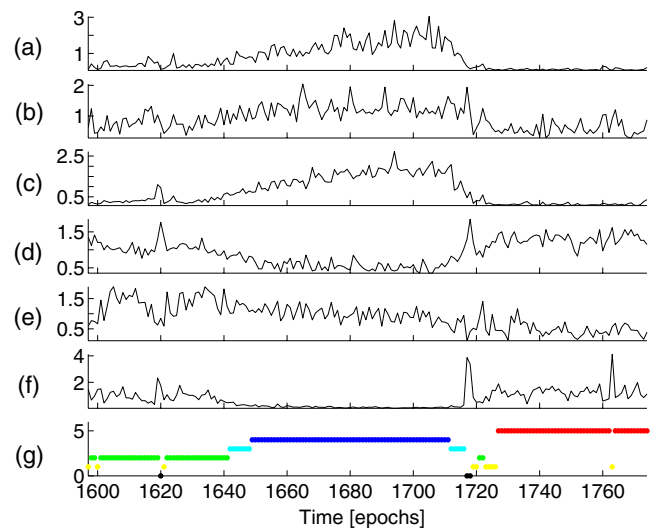


Fig. 4 Scaled features of EEG data set sc4002e0, as described in Section 4.2. The subfigures show power in the (a) delta and (b) theta bands, (c) variance, (d) spindle score, (e) maximum height of the power spectrum in the alpha band after subtraction of a linear estimate, and (f) high power fraction. Figure (g) shows the hypnogram of the EEG data, where the number and color indicate the sleep stage: awake (0, black), stage 1 (1, yellow), stage 2 (2, green), stage 3 (3, cyan), stage 4 (4, blue), and REM (5, red). The features were calculated for the data from epochs 1,597–1,774 in 30-second windows with no overlap

We then use these features as the high-dimensional input to the LLE algorithm. The 2D results for 13 nearest neighbors ($k = 13$) are displayed in Fig. 5(a). Every point in this figure represents a 30-second window of EEG data, and the color and symbol represent the sleep stage as determined by manual scoring. Here we see a very clear separation between the REM points (red circles) and those from stage 4 (blue stars), as required by our criteria for the automatic selection of the feature set. Stages 1 through 3 are located between those two groups and are arranged by sleep depth. In this example, we see a general trend of increasing sleep depth as we move to the upper right corner of the space. In addition, this low-dimensional embedding provides results with a smooth temporal progression. This is demonstrated by Fig. 5(b), where the LLE results from Fig. 5(a) are plotted versus time. In this example, the gradual transition to deep stage 4 sleep and the quick transition to REM are visible in the plot of Y_1 .

We would like to emphasize the importance of identifying strongly connected components when using LLE. Figure 6(a) shows an example of the Y_1 – Y_2 output space when LLE is performed on all 178 data points. Here, the feature set consisted of power in the delta, theta, and gamma bands, total power, maximum value of the PSD in the alpha band, and the

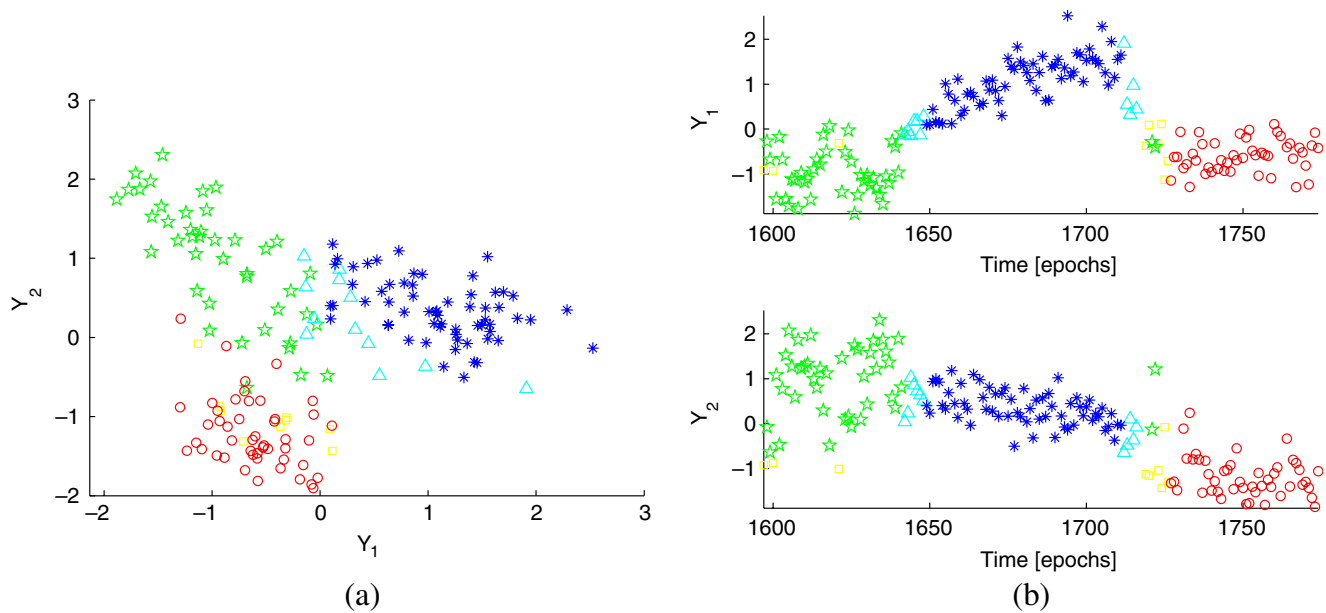


Fig. 5 (a) Results of applying LLE to EEG data using the six features in Fig. 4. The features were calculated for 30-second non-overlapping windows of data and the resulting 6-dimensional points were embedded in 2D space using LLE with $k = 13$; therefore, each point in this figure represents 30 s of EEG that has been characterized by the six features. The color and shape

indicate sleep stage based on manual scoring: awake (black +), stage 1 (yellow □), stage 2 (green ★), stage 3 (cyan △), stage 4 (blue *), and REM (red ○). (b) LLE output dimensions Y_1 and Y_2 versus time, for the results shown in (a). This demonstrates that LLE provides a low-dimensional output where the sleep state changes smoothly over time

low power fraction. Again, each point represents 30-seconds of EEG data, and the symbol (and color) are assigned based on its designated sleep stage. While there is some visible separation between the stages, the

overall trend is unclear. On the other hand, Fig. 6(b) shows the results when LLE is applied to the largest strongly connected component within the data. This component was identified as described in Section 3.2,

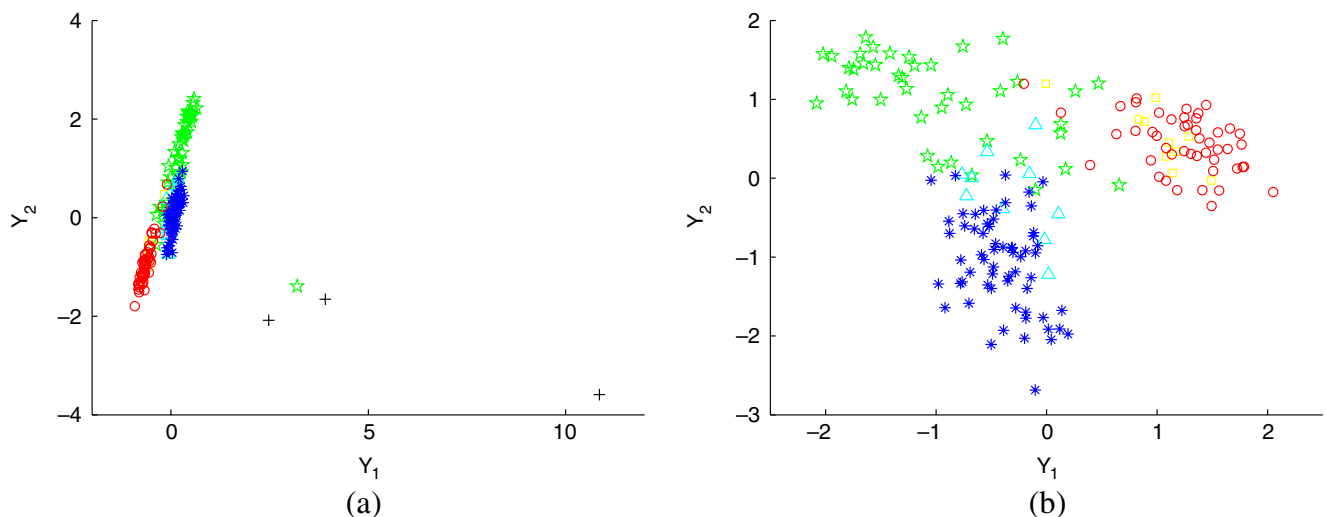


Fig. 6 LLE results on sleep EEG data before (a) and after (b) removal of eight weakly connected points. The six features were power in the delta, theta, and gamma bands, total power, maximum height of PSD above a linear estimate in the alpha band, and low power fraction, and we used $k = 13$. As before,

the color and symbol indicate sleep stage: awake (black +), stage 1 (yellow □), stage 2 (green ★), stage 3 (cyan △), stage 4 (blue *), and REM (red ○). Note the dramatic improvement in separation between sleep stages when LLE is done on only one strongly connected component in (b)

and all other points were removed before using LLE. This greatly improves the results; the data points are spread further apart, and we see a grouping of sleep stages similar to Fig. 5. Sometimes the removal of weakly connected points has a very small impact on the results, but situations like this make it a necessity. The significant improvement for this feature set allowed it to be counted as one of the “best” 11 results discovered by the automatic algorithm.

In this case, analysis of the strongly connected components resulted in the removal of eight data points:

- 3 points from waking (epochs 1,620, 1,717, and 1,718)
- 2 points from stage 2 (epochs 1,619 and 1,624)
- 2 points from stage 3 (epochs 1,712 and 1,713)
- 1 point from stage 4 (epoch 1,665)

Based on Fig. 6(a), we can see why some of these were removed; there are four points that are clearly isolated from the rest of the data. However, the removal of points from stages 3 and 4 are much less obvious. It is important to realize that, by using the concept of strongly connected components, this decision is automatic—it allows us to avoid the subjective selection of outlier points.

5 Integration of EEG data and the model sleep cycle

Thus far, we have shown that LLE is capable of distinguishing between sleep stages using only one channel of EEG and that the embedding exhibits a smooth progression over time. However, remember that our original goal was to find the relationship between EEG data and the mathematical model of the sleep cycle. Here we accomplish this by applying LLE *simultaneously* to EEG data and simulated data from the model.

5.1 Model data set

To generate the model data set, we place a grid of points on the sleep manifold (Fig. 1) and obtain the numerical solution of the cortical model at each one. We vary L over the interval $[0.5, 2]$ in increments of 0.1 and Δh_e^{rest} over $[-5, 5]$ in increments of 0.5. This gives us a total of 336 model signals for analysis; we then remove the initial transients and characterize each signal based on a subset of the features described in Section 4.2. In this way, the nonlinear sleep manifold is turned into “EEG-like” signals which are converted to high-dimensional data points for use with LLE.

For the model data set, the length of each signal is 10 s (as opposed to the 30-second windows used for the EEG data). We are able to use this shorter time because we can choose parameters in the model to simulate a stationary brain state, i.e. we can use constant values of L and Δh_e^{rest} . A test of the feature calculations for various window lengths indicated that, in many cases, the signal properties were stationary for windows greater than five seconds. Certain parts of the sleep manifold had transients lasting roughly 10 seconds.

In order to compare this model data set directly to EEG measurements, it is important that all of the basic properties match. For example, just as REM EEG signals have a much lower variance than those from stages 3 and 4, we expect that the signals from the topmost REM portion of the sleep manifold will have a smaller variance than those on the lower NREM section. However, we found that the use of a constant α , which defines the variance of the stochastic input to the model cortex in Eq. (10), does not reproduce this behavior. Therefore, we varied the value of α as we moved in the L - Δh_e^{rest} space. More specifically, we based it on the sleep manifold. Define μ_e to be a matrix of the steady-state values of h_e after they have been shifted and scaled to have a range of $[0, 1]$. Then we define a matrix of α values:

$$\alpha = \alpha \cdot (-7\mu_e + 8) . \quad (15)$$

Therefore, the REM portion of the model sleep cycle (where $\mu_e \sim 1$) will have stochastic inputs of α , while the lower NREM section (where $\mu_e \sim 0$) will have inputs of variance 8α . This stochastic input allowed us to successfully reproduce the desired range of variances in the model signals.

In addition to the variance, other features of the model data set mimic characteristics of sleep EEG. This can be verified by plotting the features as we traverse the sleep manifold. For example, power in the delta band, composite permutation entropy index (CPEI), “peak” height of the power spectral density in the alpha band, low power fraction, and high power fraction are shown in Fig. 7. The values of each feature are displayed for the grid of points in L and Δh_e^{rest} that covers the sleep manifold. For reference, the steady state values of h_e on the sleep manifold are shown in Fig. 7(a); note that this is similar to viewing Fig. 1 from the top and coloring the points based on their height. The lowest value is plotted in white and the highest value in black.

As desired, Fig. 7(b) indicates that the power in the delta band increases as the depth of sleep increases,

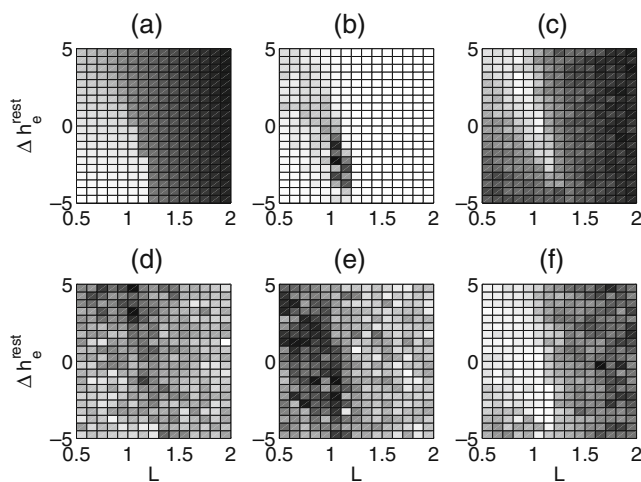


Fig. 7 Variation of five features as the surface of the sleep manifold is traversed in L - Δh_e^{rest} space. Each feature has been scaled by its RMS value and depicted in *grayscale*, with *white* indicating the lowest values and *black* representing the highest values. **(a)** The steady state values of h_e from the sleep manifold in Fig. 1. The *black points* represent the upper REM branch, the *white points* represent NREM, and the fold is located at roughly $L = 1.2$. The other subfigures show **(b)** power in the delta band, **(c)** permutation entropy, **(d)** maximum height of PSD above a linear estimate in the alpha band, **(e)** low power fraction, and **(f)** high power fraction. These five features use α as defined in Eq. (15). They show that the representation of REM and NREM in the model is consistent with the characteristics of sleep EEG

with the largest values occurring near the quick transition to REM sleep. Similarly, Fig. 7(e) and (f) show that the fraction of power in the low frequencies is greater during NREM sleep, while the fraction of power at high frequencies is greater during REM sleep. Consistent with previous reports that the CPEI decreases with depth of anesthesia (Olofson et al. 2008), we see in Fig. 7(c) that the CPEI decreases with sleep depth in the model. Figure 7(d) shows that the region of greatest alpha power is located in the upper left corner, for small values of L and large values of Δh_e^{rest} . As a means of comparison, the same five features were applied to a sample of EEG data and are displayed in Fig. 8.

5.2 Application of LLE to a hybrid data set

We now join the EEG measurements and the model data into one hybrid data set and use it as an input to the LLE algorithm. This simultaneously finds the low-dimensional embedding for both data types and allows us to infer a correspondence between them. For example, Fig. 9(a) shows the result of applying LLE to the grid of 336 model points and a full night's sleep from EEG data set sc4002e0 (epochs 800 to 2,000).

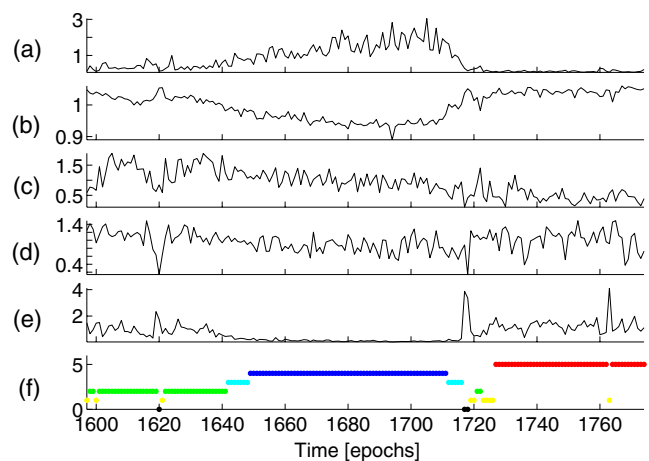


Fig. 8 Variation of the features from Fig. 7 when they are applied to sleep EEG data, rather than model data. The sample of EEG data was taken from sc4002e0, and each feature has been scaled by its RMS value. The subfigures show **(a)** power in the delta band, **(b)** permutation entropy, **(c)** maximum height of PSD above a linear estimate in the alpha band, **(d)** low power fraction, **(e)** high power fraction, and **(f)** hypnogram of the EEG data. The colors and numbering for the hypnogram are the same as those used for Fig. 4. Note that the values of these features (relative to sleep stage) are consistent with the model results in Fig. 7

The input data was composed of the five features from Fig. 7: power in the delta band, CPEI, maximum height of the PSD in the alpha band (relative to a linear estimate), and the low and high power fractions. We used $k = 14$, and only three points were removed by analysis of the strongly connected components.

In Fig. 9(a), the model data is represented by dots, where the color denotes the steady-state value of h_e associated with that point; in general, the red points represent the REM portion of the manifold, while the blue points represent NREM. On the other hand, the sleep EEG data points are rings, where the color is chosen based on sleep stage. Note that, for clarity, only the first 500 EEG data points were included in the figure.

The most important aspect of this result is that the EEG data points and model points overlap each other in the Y_1 - Y_2 output space. This implies that model points have EEG data points as nearest neighbors (and vice versa) and verifies that LLE has associated the two data types with one another. Without fidelity of the model and careful choice of EEG features, we would have likely obtained a result with one cluster of EEG points and a completely separate cluster of model points. Further, LLE appears to have matched the sleep stages between the two data types—the deepest sleep (blue for both EEG and model) appears in the lower left corner, and REM (red) is embedded in a vertical

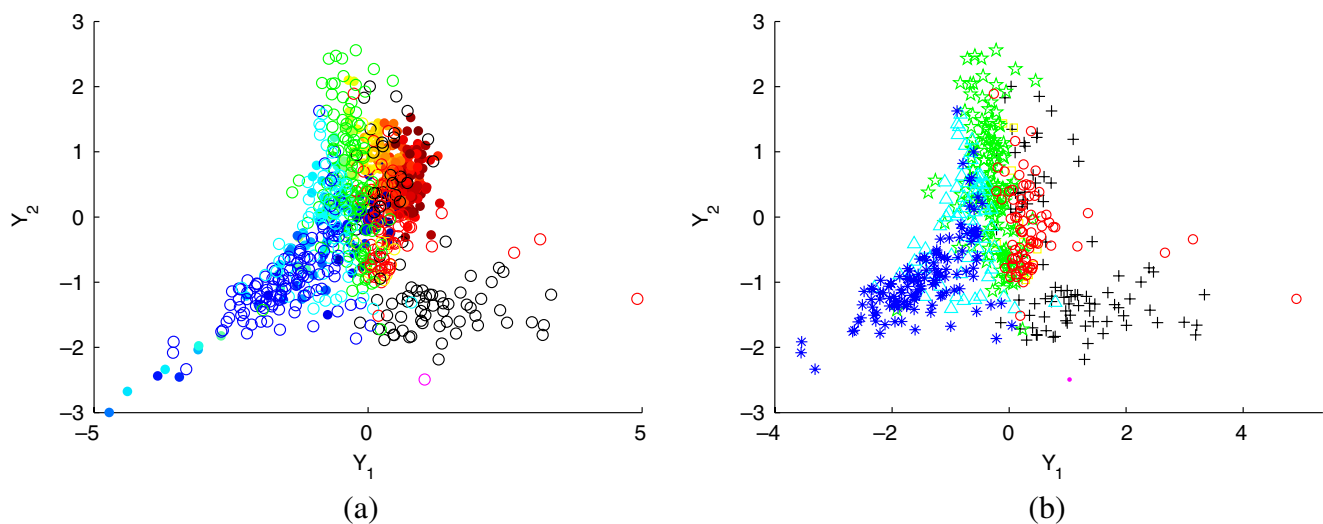


Fig. 9 (a) LLE results for a hybrid data set containing both sleep EEG data and numerical solutions of the cortical model. We used the five features from Fig. 7 and set $k = 14$. The rings represent EEG data and are colored by sleep stage. While the analysis included 1,200 windows of EEG data, only 500 are displayed here for clarity. The solid dots represent data from the model; they are colored based on the mean value of h_e at that point, where

red represents the highest (REM) values, and *dark blue* marks the lowest (NREM) values. Note that the data and model points overlap in the output space and that the arrangement of sleep stages is very similar. (b) LLE results showing the EEG data only, using the same colors and symbols as Fig. 6. This allows us to see that the data has been roughly separated by sleep stage

band where Y_1 is in the range $[-1, 0]$. The separation between sleep stages can be seen more clearly in Fig. 9(b), which displays only the EEG data points from Fig. 9(a). Here we see that the stages are grouped; even the REM points and the awake points are separated, despite the fact that their EEG traces are characterized by very similar features. If we were to plot the Y_1 and Y_2 values of the EEG data points as they evolve in time, we would see a very similar result to the one in Fig. 5(b). Here, the Y_1 direction appears to be an approximate indicator of sleep depth.

5.3 Connection to the theoretical sleep manifold

So far, we have seen that LLE provides a qualitatively similar embedding for REM and NREM points in both EEG measurements and simulated model data. However, we would like to quantify this relationship. In other words, we would like to associate each EEG data point with a position on the sleep manifold in the $L-\Delta h_e^{rest}$ space. This will allow us to infer the model trajectory of a subject's actual brain state as it moves along the manifold.

To do this, we use the results in Fig. 9(a) and again turn to the concept of nearest neighbors. Using $k = 14$, we calculate the nearest neighbors of every point in the Y_1 – Y_2 space. We then identify *model* points that are nearest neighbors of *EEG* data points. Each one of those model points has an associated position on the

sleep manifold; we assume that the $L-\Delta h_e^{rest}$ positions of the model nearest neighbors will be the most closely associated positions for the EEG data point.

We can visualize this concept by creating histograms of the model nearest neighbors and separating them by sleep stage (Fig. 10(a)). Every time a model point is a nearest neighbor of an EEG point, we increment the count at the model point's associated location in $L-\Delta h_e^{rest}$ for the sleep stage of the EEG point. We then create grayscale plots of the total counts, where white indicates that a location was never a nearest neighbor of that sleep stage and black indicates that it was a nearest neighbor many times.

For example, (i)–(vi) in Fig. 10(a) correspond to awake, REM, and stages 1–4, respectively. The thick vertical line at $L = 1.2$ marks the approximate location of the fold. As we move from REM to the deeper stages of sleep, we can see a continuous progression along the sleep manifold. In this example, REM and stage 1 sleep generally associate themselves with locations on the right half of the manifold (and a small piece of the lower left corner). Then in stage 2 sleep, we move to the left half of the manifold; here, we see two distinct groups of points, with a majority landing in the group that borders the area associated with REM and Stage 1. Stage 3 is associated with a cluster of points starting in the upper left-hand corner and approaching the fold. Stage 4 continues this progression and is located in a band of points leading up to the fold.

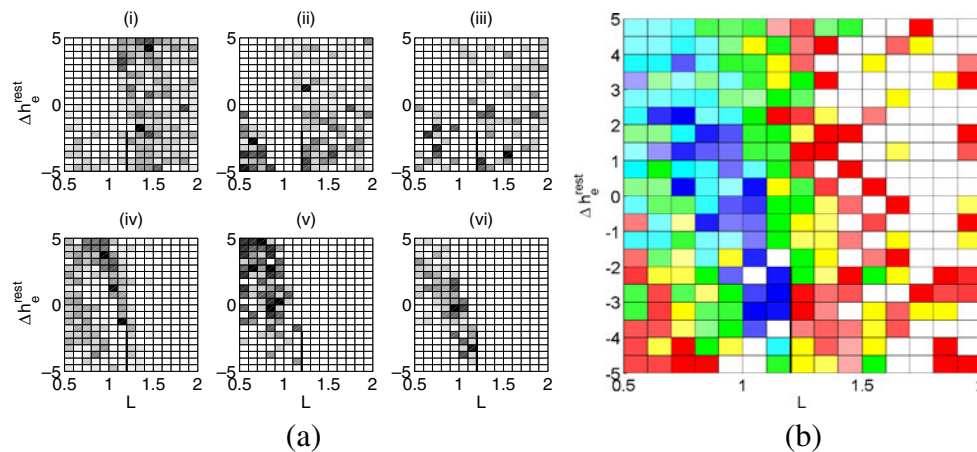


Fig. 10 Association between EEG data set sc4002e0 and the sleep manifold. Each picture shows the sleep manifold in L - Δh_e^{rest} space, with a *heavy black line* to indicate the location of the fold. **(a)** Histograms of nearest neighbors for (i) waking, (ii) REM, (iii) stage 1, (iv) stage 2, (v) stage 3, and (vi) stage 4 sleep. The shading of each square indicates the number of times that location on the sleep manifold was a nearest neighbor of EEG data in that stage. For example, (vi) shows that stage 4 sleep most often associates itself with the lower NREM branch of solutions

leading up to the fold. **(b)** A composite picture of the results in (a), where each location is colored based on the sleep stage with the most neighbors at that point, relative to the total number of neighbors associated with that stage. Again, we use stage 1 (yellow), stage 2 (green), stage 3 (cyan), stage 4 (blue), and REM (red). The intensity of color is scaled based on the percentage of neighbors that come from that stage; the more saturated the color, the greater the percentage. Waking points were excluded

We can then create a composite plot that combines all five sleep stages. We neglect the waking points for this task because the current model does not effectively distinguish between the waking and REM states, although this is certainly an issue that may be addressed in the future. For every location on the manifold, we determine which sleep stage it was most closely associated with and color it accordingly. To do this, we scale the number of nearest neighbors for each stage by the total number for that stage; then, for every position on the manifold, we choose the stage with the highest value. This accounts for the fact that the subjects do not spend an equal amount of time in each sleep stage (otherwise, more time spent in a certain stage would lead to more nearest neighbors and a greater likelihood of dominating this composite plot). As in previous figures, we use red for REM, yellow for stage 1, green for stage 2, cyan for stage 3, and blue for stage 4. The intensity of the color is assigned based on the percentage of times it was associated with that sleep stage. Suppose a certain point on the manifold was a neighbor of stage 2 twelve times, a neighbor of stage 1 five times, and a neighbor of REM three times. We would color that point green to indicate stage 2 sleep, and its saturation value would be $12/(12 + 5 + 3) = 0.6$. In other words, the intensity of the color is a “confidence” measure; the more saturated the color, the more closely it is associated with that sleep stage. The composite figure for the data in Fig. 10(a) is shown in Fig. 10(b).

5.4 Inclusion of additional data sets

It is important that this method of analysis works consistently for different subjects with a variety of sleeping patterns. We tested this capability using the full night of sleep from each of the remaining three data sets: sc4012e0, sc4102e0, and sc4112e0. Rather than start from scratch and re-run the LLE algorithm, we projected the new data onto the existing embedding. For a new input \mathbf{x} , this is a three-step process (Saul et al. 2003):

1. Find the k nearest neighbors of each new data point among the points in the existing embedding.
2. Compute the best linear reconstruction w_j of each new point using only its nearest neighbors. Again, we enforce the constraint that the weights used in the reconstruction sum to one: $\sum_j w_j = 1$.
3. Calculate the output for the new data points: $\mathbf{y} = \sum_j w_j \mathbf{Y}_j$, where \mathbf{Y} contains the original embedding coordinates and j cycles through the neighbors of \mathbf{x} .

This is more computationally efficient than running the entire algorithm again, and it guarantees that the output embedding will not change as we add new data. Most importantly, this makes it possible to do continuous real-time monitoring of EEG data; a new point could be projected onto the results every 30 s (or less) as the subject sleeps.

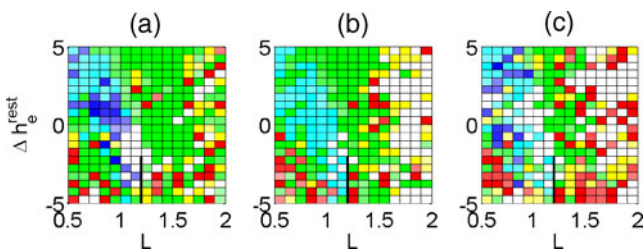


Fig. 11 Composite plots for EEG data sets (a) sc4012e0, (b) sc4102e0, and (c) sc4112e0, when they are projected onto the LLE results from Fig. 6(b), as described in Section 5.4. These pictures are analogous to Fig. 10(b) and use the same color scheme. Note that the results are consistent with those for sc4002e0 in Fig. 10(b); over various subjects, the sleep stages are generally associated with the same positions on the sleep manifold

When we project the sleep data from files sc4012e0, sc4102e0, and sc4112e0 onto the embedding derived from sc4002e0, we obtain the composite pictures in Fig. 11(a)–(c), respectively. All three results are consistent with one another, despite coming from different subjects and containing a minimal amount of stage 3 and 4 deep sleep. The only exception to this is stage 4 sleep in Fig. 11(c); however, it is important to note that only 21 points out of 1100 were denoted as stage 4 sleep for this subject, and those points were not all consecutive. Therefore, the subject had only transient movements into stage 4 from stage 3, and it is perhaps not surprising that the results show the stage 4 EEG points mixed in with those from stage 3. Also note that

the placement of the sleep stages in Fig. 11 is consistent with the results in Fig. 10.

Lastly, we combine the results from all four data sets (the original embedding with sc4002e0 plus three projected data sets) to produce Fig. 12. The histograms in Fig. 12(a) were created by a simple summation of the nearest neighbor histograms for all four data sets. The composite plot in Fig. 12(b) was then generated according to the logic described in Section 5.3 using the combined histogram data. In all, these results are based on almost 40 h of EEG data from four different subjects. Again, they are consistent with the individual results and they show a clear picture of the sleep manifold regions associated with each sleep stage. It is also noteworthy that only a handful of points on the sleep manifold (colored white in the composite picture) were never nearest neighbors of an EEG data point.

This picture may be very useful in the analysis of seizures during sleep. Imagine taking another new sleep EEG data set, this time from an epileptic subject, and projecting it onto these results. By following the location in $L-\Delta h_e^{\text{rest}}$ as the subject sleeps, we can get an idea of the sleep stage as it is traditionally defined, and we can also identify that stage in more detail and detect nearness to transitions between stages. The grid of points on the sleep manifold essentially gives us descriptions of 336 different brain states associated with sleep. We expect that future research will identify the locations on the sleep manifold where seizures are most likely to occur. With that knowledge, if the sleep state

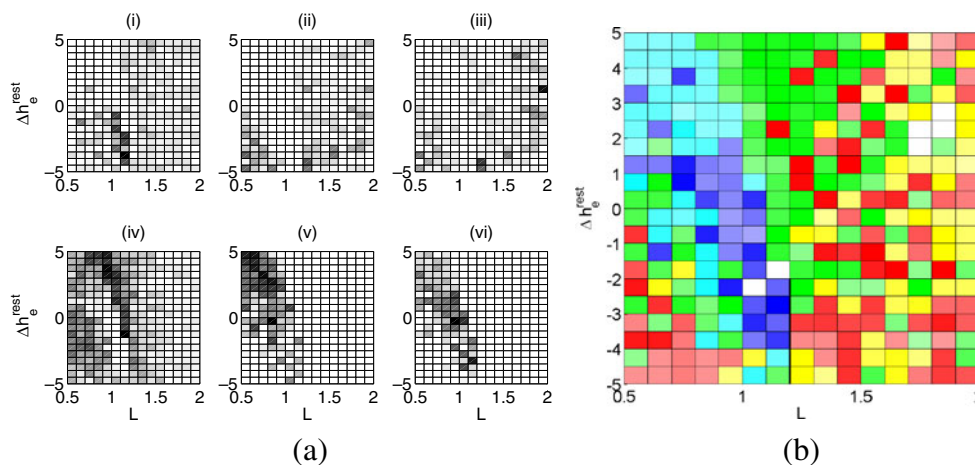


Fig. 12 Combined association between all four EEG data sets and the sleep manifold. The data set sc4002e0 was directly compared to the cortical model using LLE, and the remaining three data sets were projected onto those results as described in Section 5.4. (a) Total histograms of nearest neighbors, separated by sleep stage; these were calculated by summing the histograms from all four EEG data sets. The pictures show awake, REM,

and stages 1–4 in (i) through (vi), respectively. (b) The total composite picture for all four data sets. This was generated from the histogram data in (a) and is analogous to Figs. 10(b) and 11. Again, this is consistent with previous results and shows the regions of the sleep manifold most closely associated with each sleep stage

characterization is done continuously while the subject is sleeping, this may allow for the prediction (and possibly prevention) of seizures.

We emphasize the fact that the coloring in the composite pictures (Figs. 10(b), 11, and 12(b)) is based on the subjective scoring of sleep data. The reliability of categorizing individual epochs of data has been reported at 73% for scorers from different labs (Norman et al. 2000) and as high as 90% for scorers from the same lab (Whitney et al. 1998). It has also been shown that reliability varies by sleep stage, with stage 2 having the highest level of agreement between scorers (78.3%) and stage 1 having the lowest (41.8%) (Norman et al. 2000). This certainly affects our results. For example, imagine if some of the points scored as REM that landed in the range $1.6 < L < 2$ on the sleep manifold were instead scored as stage 1. Then the right side of the composite picture would be completely yellow and the region associated with stage 1 would be more clear. Therefore, the composite pictures should be seen as “guides” to tie the analysis back to the traditional definitions of the sleep stages, not as the ultimate truth. As mentioned in the previous paragraph, we are most interested in the *position* on the sleep manifold, the trajectory that results as the subject sleeps, and the relationship of this trajectory to the regions where seizures may be most likely to occur.

6 Summary

Mathematical models represent an opportunity for exploration and prediction. In this case, a model of the human sleep cycle creates the possibility for a more detailed description of sleep states, with application to the prediction and analysis of seizures during sleep. The first step in such an endeavor is always to connect the model to the real world through experimental data.

Here we have used locally linear embedding to directly associate human sleep EEG data with the mathematical model. We first showed that LLE has the ability to distinguish between sleep stages when applied to EEG data alone. This analysis can reliably separate REM and NREM sleep data and provide a smooth temporal progression through the various stages of sleep. We also presented the concept of strongly connected components as a method of automatic outlier rejection for EEG data and discussed a method for the selection of EEG features used in the analysis. Then, by using LLE on a hybrid data set containing both sleep EEG and signals generated from the mathematical sleep cycle, we were able to quantify the relationship between the model and the data. This enabled us to take any

sample of sleep EEG data and associate it with a position among the continuous range of sleep states provided by the model. In addition, this approach yields consistent results for various subjects over a full night of sleep and can be done online as the subject sleeps. This suggests a wide range of possibilities for future investigation.

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A model of feedback control for the charge-balanced suppression of epileptic seizures

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Abstract Here we present several refinements to a model of feedback control for the suppression of epileptic seizures. We utilize a stochastic partial differential equation (SPDE) model of the human cortex. First, we verify the strong convergence of numerical solutions to this model, paying special attention to the sharp spatial changes that occur at electrode edges. This allows us to choose appropriate step sizes for our simulations; because the spatial step size must be small relative to the size of an electrode in order to resolve its electrical behavior, we are able to include a more detailed electrode profile in the simulation. Then, based on evidence that the mean soma potential is not the variable most closely related to the measurement of a cortical surface electrode, we develop a new model for this. The model is based on the currents flowing in the cortex and is used for a simulation of feedback control. The simulation utilizes a new control algorithm incorporating the total integral of the applied electrical potential. Not only does this succeed in suppressing the seizure-like oscillations, but it guarantees that the applied signal will be charge-balanced and therefore unlikely to cause cortical damage.

Keywords Epilepsy · Seizure · Feedback · Control · Electrocorticogram · Human

1 Introduction

The recurrent, unprovoked seizures associated with epilepsy can have a devastating effect on those with this disorder. Basic parts of every day life such as driving and obtaining employment become very difficult. While many people with epilepsy can control their seizures with medication, roughly thirty percent do not respond to this type of treatment and therefore seek out alternatives such as surgery (The Epilepsy Foundation 2009). The surgical procedure involves resecting the seizing portion of the brain, often part of the cortex, while avoiding any areas that provide vital functions such as speech, memory, and vision (The Epilepsy Foundation 2009). Because this is a very invasive procedure that does not guarantee success, other alternatives are being investigated. One of these is automatic feedback control, where subdural electrodes on the cortical surface would detect the seizure and apply an electrical signal to disrupt the abnormal electrocorticogram (ECoG) activity.

This method of treatment is currently being studied by experimentalists. It has been shown that the application of electric fields to rat cortex *in vitro* can modulate the behavior of seizure-like waves (Richardson et al. 2005). *In vivo* experiments on rats demonstrated that stimulation via proportional feedback can temporarily suppress seizure activity (Gluckman et al. 2001). A subsequent set of experiments showed that an increase in the amplitude of the proportional control feedback gain corresponds to decreases in both seizure amplitude

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(measured as a reduction in the amplitude variance) and Teager energy (Colpan et al. 2007). The Teager energy can be reduced by a decrease in the amplitude of a signal or a lowering of the frequencies in its power spectrum.

While these experiments suggest further exploration, they are all restricted to animal implementation. Therefore, to gain insight into the feasibility of human use, we turn first to mathematical models. The stochastic partial differential equation (SPDE) model of the cortex used here can support seizure-like oscillations that are qualitatively and quantitatively similar in frequency of maximum power and propagation speed to those seen in humans with epilepsy (Kramer et al. 2005). We previously demonstrated that various methods of feedback control can suppress these simulated seizures (Kramer et al. 2006), and we added the capability of looking at spatial properties of feedback control, such as electrode size and spacing (Lopour and Szeri 2008).

Our aim in the present work is to make this model more biologically relevant by refining the representation of feedback control. This will facilitate future comparison with experimental data. There are four key improvements in our approach that are presented in this article:

1. We verify the strong convergence of the numerical solution to the SPDE model.
2. Based on the studies of convergence, we utilize a smaller step size in our simulations, thereby allowing the inclusion of a more detailed electrode profile.
3. We develop a better motivated model of the signal measured by an electrode on the cortical surface. This model is used to calculate the applied electrical signal for feedback control.
4. Feedback control is performed with a new algorithm incorporating an integral component. This ensures that the applied signal is charge-balanced, which is thought to minimize damage of cortical tissue.

The organization of the paper is as follows. We first briefly review the SPDE cortical model that will be used in our simulations (Section 2), and then we discuss the convergence of its numerical solutions (Section 3). Next, we present the new model for electrode measurements (Section 4). Finally, we incorporate these improvements into simulations of feedback control, while implementing a new integral control law (Section 5).

2 Cortical model

To model the electrical activity of the human cortex, including seizure waves, we choose a set of stochastic PDEs that has been developed and adapted over the past decade (Liley et al. 1999, 2002; Steyn-Ross et al. 2003). The mesoscale nature of this model makes it well-suited to EEG-based applications such as epilepsy, sleep (Wilson et al. 2006), and anesthesia (Steyn-Ross et al. 2004) because it is based on length scales similar to commercial electrode arrays. It is a mean-field model, meaning that all of its variables represent spatially averaged properties of populations of neurons. This is similar to the manner in which an electrode provides a measurement based on the collective behavior of many neurons.

In 2006, the equations were restated in a dimensionless form by Kramer et al. (2007). This is the formulation of the model we will use here; it is a system of eight coupled nonlinear PDEs with stochastic inputs:

$$\frac{\partial \tilde{h}_e}{\partial \tilde{t}} = 1 - \tilde{h}_e + \Gamma_e(h_e^0 - \tilde{h}_e)\tilde{I}_{ee} + \Gamma_i(h_i^0 - \tilde{h}_e)\tilde{I}_{ie} + u \quad (1)$$

$$\frac{\partial \tilde{h}_i}{\partial \tilde{t}} = 1 - \tilde{h}_i + \Gamma_e(h_e^0 - \tilde{h}_i)\tilde{I}_{ei} + \Gamma_i(h_i^0 - \tilde{h}_i)\tilde{I}_{ii} \quad (2)$$

$$\left(\frac{1}{T_e} \frac{\partial}{\partial \tilde{t}} + 1\right)^2 \tilde{I}_{ee} = N_e^\beta \tilde{S}_e[\tilde{h}_e] + \tilde{\phi}_e + P_{ee} + \tilde{\Gamma}_1 \quad (3)$$

$$\left(\frac{1}{T_e} \frac{\partial}{\partial \tilde{t}} + 1\right)^2 \tilde{I}_{ei} = N_e^\beta \tilde{S}_e[\tilde{h}_e] + \tilde{\phi}_i + P_{ei} + \tilde{\Gamma}_2 \quad (4)$$

$$\left(\frac{1}{T_i} \frac{\partial}{\partial \tilde{t}} + 1\right)^2 \tilde{I}_{ie} = N_i^\beta \tilde{S}_i[\tilde{h}_i] + P_{ie} + \tilde{\Gamma}_3 \quad (5)$$

$$\left(\frac{1}{T_i} \frac{\partial}{\partial \tilde{t}} + 1\right)^2 \tilde{I}_{ii} = N_i^\beta \tilde{S}_i[\tilde{h}_i] + P_{ii} + \tilde{\Gamma}_4 \quad (6)$$

$$\left(\frac{1}{\lambda_e} \frac{\partial}{\partial \tilde{t}} + 1\right)^2 \tilde{\phi}_e = \frac{1}{\lambda_e^2} \frac{\partial^2 \tilde{\phi}_e}{\partial \tilde{x}^2} + \left(\frac{1}{\lambda_e} \frac{\partial}{\partial \tilde{t}} + 1\right) N_e^\alpha \tilde{S}_e[\tilde{h}_e] \quad (7)$$

$$\left(\frac{1}{\lambda_i} \frac{\partial}{\partial \tilde{t}} + 1\right)^2 \tilde{\phi}_i = \frac{1}{\lambda_i^2} \frac{\partial^2 \tilde{\phi}_i}{\partial \tilde{x}^2} + \left(\frac{1}{\lambda_i} \frac{\partial}{\partial \tilde{t}} + 1\right) N_i^\alpha \tilde{S}_i[\tilde{h}_i] \quad (8)$$

All variables are dimensionless and are functions of time (\tilde{t}) and one spatial dimension (\tilde{x}). The \tilde{h} state variable is the mean soma potential for a neuronal population, while \tilde{I} represents postsynaptic activation due to local, long-range, and subcortical inputs, and $\tilde{\phi}$ is a long-range (corticocortical) input. The subscripts e and i denote affiliation with the excitatory and inhibitory neuron populations, respectively; variables with two subscripts represent the transmission of information from one population to another, e.g. \tilde{I}_{ie} is the postsynaptic activation of the excitatory population due to

Table 1 Dimensionless variables and parameters of the SPDE cortical model

Symbol	Definition	Description	Typical value
$\tilde{h}_{e,i}$	$h_{e,i}/h^{rest}$	Spatially averaged soma potential for neuron populations	–
$\tilde{I}_{ee,ei}$	$I_{ee,ei}\gamma_e/(G_e \exp(1)S^{\max})$	Postsynaptic activation due to excitatory inputs	–
$\tilde{I}_{ie,ii}$	$I_{ie,ii}\gamma_i/(G_i \exp(1)S^{\max})$	Postsynaptic activation due to inhibitory inputs	–
$\tilde{\phi}_{e,i}$	$\phi_{e,i}/S^{\max}$	Long-range (corticocortical) input to e and i populations	–
\tilde{t}	t/τ	Time (dimensionless)	–
\tilde{x}	$x/(\tau\tilde{v})$	Space (dimensionless)	–
$\Gamma_{e,i}$	$\frac{G_{e,i} \exp(1)S^{\max}}{\gamma_{e,i} h_{e,i}^{rev} - h^{rest} }$	Influence of synaptic input on mean soma potential	1.42×10^{-3} , 0.0774
$h_{e,i}^0$	$h_{e,i}^{rev}/h^{rest}$	Reversal potential	–0.643, 1.29
$T_{e,i}$	$\tau\gamma_{e,i}$	Neurotransmitter rate constant	12.0, 2.6
$\lambda_{e,i}$	$\tau\tilde{v}\Lambda_{ee,ei}$	Inverse length scale for corticocortical connections	11.2, 18.2
$P_{ee,ei}$	$p_{ee,ei}/S^{\max}$	Subcortical input from excitatory population	11.0, 16.0
$P_{ie,ii}$	$p_{ie,ii}/S^{\max}$	Subcortical input from inhibitory population	16.0, 11.0
$N_{e,i}^{\alpha}$	–	Number of distant (corticocortical) connections from excitatory populations to e and i populations	4,000, 2,000
$N_{e,i}^{\beta}$	–	Number of local synaptic connections from e and i populations	3,034, 536
$\tilde{g}_{e,i}$	$g_{e,i}h^{rest}$	Slope at inflection point of sigmoid function \tilde{S}_e	–19.6, –9.8
$\tilde{\theta}_{e,i}$	$\theta_{e,i}/h^{rest}$	Inflection point for sigmoid function \tilde{S}_e	0.857, 0.857

Values were taken from Kramer et al. (2007). For descriptions and values of the dimensional variables please refer to Steyn-Ross et al. (2003)

inputs from the inhibitory population. In Eq. (1) we have added the variable u to represent the signal applied by a cortical surface electrode for feedback control. This will be discussed further in Section 5. For descriptions of all model variables and parameters, please refer to Table 1.

To appreciate the model as a whole, let us first look at the equations governing the excitatory neuron population, depicted graphically in Fig. 1. Equation (1) for \tilde{h}_e is reminiscent of the leaky integrate-and-fire model of a neuron, where the derivative of the membrane potential equals the resting potential minus the membrane potential plus any existing current inputs

(Dayan and Abbott 2001). Here, the resting potential is “1” due to the dimensionless nature of the system. The inputs are \tilde{I}_{ee} and \tilde{I}_{ie} , which evolve according to (3) and (5), respectively, based on three types of synaptic input: local, long-range, and subcortical.

– **Local inputs**, such as those from within the same macrocolumn, are represented by terms of the form $N_e^{\beta}\tilde{S}_e[\tilde{h}_e]$, where \tilde{S}_e is a dimensionless sigmoid function:

$$\tilde{S}_e[\tilde{h}_e] = \frac{1}{1 + \exp[-\tilde{g}_e(\tilde{h}_e - \tilde{\theta}_e)]} \quad (9)$$

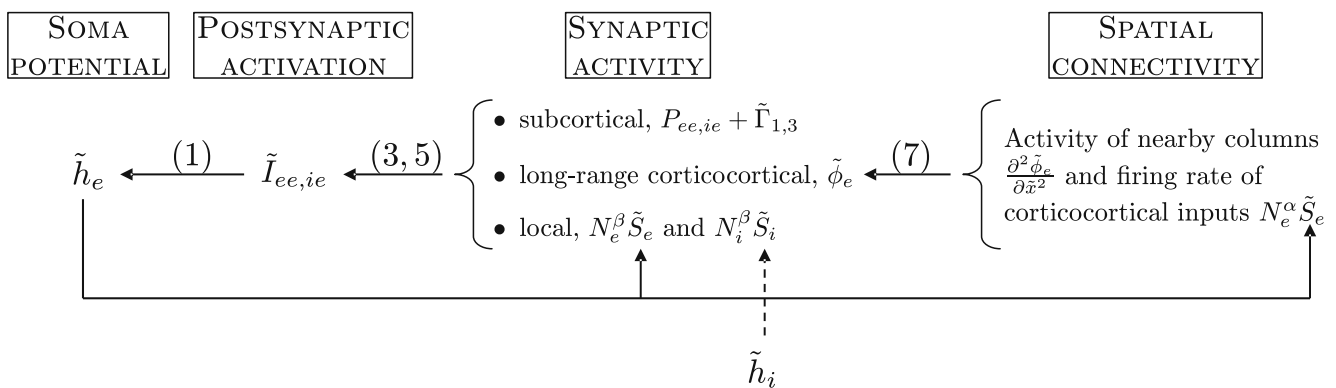


Fig. 1 Flow chart representation of model Eqs. (1), (3), (5), and (7), which govern the excitatory population. The boxes describe the physiological significance of the model variables and parameters listed beneath them. Note that feedback occurs through the

sigmoid function \tilde{S}_e , which is a function of \tilde{h}_e . This population is also coupled to the inhibitory population through local inputs described by $\tilde{S}_i[\tilde{h}_i]$. For a cell-based depiction of the model, see Steyn-Ross et al. (1999)

This converts the mean soma potential of the excitatory population to its mean firing rate.

- **Long-range inputs** represent signals from other cortical macrocolumns and are defined by $\tilde{\phi}_e$. The behavior of this variable is governed by (7). Note the similarity of this equation to the standard PDE wave equation; the idea that cortical tissue can support wave propagation is central to our simulation of epileptic seizures.
- **Subcortical inputs** are predominantly from the thalamus and contain both constant (P_{ee}) and stochastic ($\tilde{\Gamma}_1$) parameters. We define the stochastic term by

$$\tilde{\Gamma}_1 = \alpha_{ee} \sqrt{P_{ee}} \xi_1[\tilde{x}, \tilde{t}], \quad (10)$$

where α_{ee} is a constant and ξ_1 is zero mean, Gaussian white noise in time and one spatial dimension. When the SPDEs are solved numerically, the cumulative effect of this stochastic process will be Brownian motion. To ensure that the properties of this signal remain constant regardless of step size, we scale the discrete randomly generated numbers $R(m, n)$ by the simulation time step:

$$\xi_1[\tilde{x}, \tilde{t}] = \frac{R(m, n)}{\sqrt{\Delta \tilde{t}}}. \quad (11)$$

The variables m and n are indices of space and time, so a single point is represented by spatial position $\tilde{x} = m\Delta\tilde{x}$ at time $\tilde{t} = n\Delta\tilde{t}$. Note that the discrete update form of (11) is

$$\xi_1[\tilde{x}, \tilde{t}]\Delta\tilde{t} = \sqrt{\Delta\tilde{t}} R(m, n), \quad (12)$$

which will be used in all numerical experiments.

Thus, Eqs. (1), (3), (5), and (7) govern the excitatory population, while the remaining equations represent the inhibitory population and have exactly the same form. Together they compose the full cortical model.

There are several parameters that are especially relevant to the following numerical studies. The parameter P_{ee} represents input from the population of subcortical excitatory neurons (such as those in the thalamus), and Γ_e denotes the influence of synaptic input on the mean soma potential. Changes in these parameters allow for transition between normal cortical function ($P_{ee} = 11.0$ and $\Gamma_e = 1.42 \times 10^{-3}$) and the hyperexcited “seizure” state of the SPDE model (say, $P_{ee} = 548.0$ and $\Gamma_e = 0.8 \times 10^{-3}$). At low levels of excitation corresponding to low levels of P_{ee} , the mean soma potential of the excitatory neurons \tilde{h}_e produces random fluctuations similar to those seen in an EEG measurement. However, at increased levels of subcortical excitation, the simulated cortex develops large amplitude seizure-like

oscillations. Our goal is to suppress this pathological behavior via feedback control consisting of measurements from the cortical surface and the application of a potential based on those measurements.

3 Strong convergence of numerical solutions

Before performing simulations of feedback control, we must ensure that we can obtain accurate numerical solutions to this system of SPDEs. We will use a predictor-corrector algorithm written in MATLAB, so the accuracy of the solution will be determined by our choice of step sizes in space and time. In addition to considering the system equations and solution method, we shall assume that a typical cortical surface electrode is of order 1 cm in diameter. While the previously used step size of 7 mm (Lopour and Szeri 2008; Kramer et al. 2006) may have accurately solved the differential equations, it was not small enough for sufficient spatial resolution of the behavior of the cortical tissue underneath the electrodes. We will use much smaller step sizes in order to achieve both of these objectives.

3.1 Method

To determine the magnitude of these step sizes, we will examine the strong convergence of solutions to the cortical model. This will be accomplished by generating *equivalent* Brownian paths at several step sizes and demonstrating that the solutions converge as the step size decreases (note that this differs from weak convergence, which looks at the expected value of the solution over all possible Brownian paths) (Higham 2001). This task will be complicated by the fact that both the stochastic inputs and the numerical solutions vary in space and time.

Recall from (12) that the grid of stochastic inputs is defined by R , which consists of M independent Brownian paths, each of length N . This corresponds to M points in space at a reference step size of $\Delta x = \Delta x_0$ and N points in time at a reference step size of $\Delta t = \Delta t_0$. Therefore, we denote individual points by $R(m, n)$, $\forall m = 1, \dots, M$ and $\forall n = 1, \dots, N$. Then we can represent the same Brownian paths at a coarser step size $2\Delta t_0$ by adding together every two adjacent elements in time:

$$\tilde{R}(m, n) = R(m, 2n - 1) + R(m, 2n), \quad (13)$$

where $n = 1, \dots, N/2$ (Gaines 1995). We do not need any special scaling factors here because this combination of neighboring terms is consistent with the

definition of a Brownian path. Similarly, we can represent the stochastic input at step size $2\Delta x_0$ by adding together adjacent elements in space and scaling to keep the variance constant (Gaines and Lyons 1997):

$$\tilde{R}(m, n) = \frac{1}{\sqrt{2}} (R(2m-1, n) + R(2m, n)), \quad (14)$$

where $m = 1, \dots, M/2$. The factor $1/\sqrt{2}$ is necessary because the stochastic inputs are independent in the spatial direction; it will be used for the *relative* scaling of the inputs at different step sizes for the purpose of determining convergence, but will not be present in a typical simulation of feedback control.

Now we can directly compare numerical solutions at decreasing step sizes (e.g. $\Delta x = 4\Delta x_0, 2\Delta x_0, \Delta x_0$) under equivalent stochastic inputs. We want to verify that the solution converges as we approach Δx_0 .

3.2 Results

First, we look at the convergence in time using the method described above. We remove the spatial terms from the cortical model to reduce it to an ODE and then perform simulations with decreasing values of the time step. These indicate that the solution converges around $\Delta t = 5 \times 10^{-4}$ s (Fig. 2). The two smallest time steps in the figure, $\Delta t = 5 \times 10^{-4}$ and 2.5×10^{-4} s give very similar results for h_e .

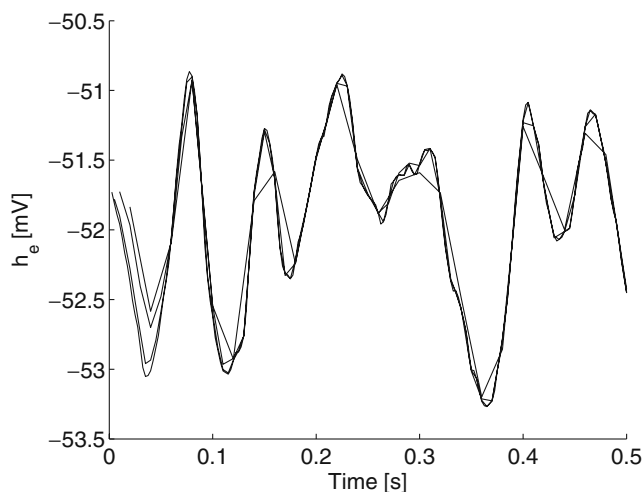


Fig. 2 Convergence of numerical solutions as the time step is decreased using the method described in Section 3.1. Here the spatial terms in the model have been removed to reduce it to an ODE. We used $\Delta t_0 = 2.5 \times 10^{-4}$ s, and we plotted the solutions for $\Delta t = 8\Delta t_0, 4\Delta t_0, 2\Delta t_0$, and Δt_0 . The two smallest time steps give overlapping results, indicating that the solution has converged. This study was done with typical excitation $P_{ee} = 11.0$ and $\Gamma_e = 1.42 \times 10^{-3}$

Next, we study the convergence in space. We begin with a Δx that is smaller than 1 cm because we desire to resolve the solution across an electrode. As Δx decreases, we see that the accuracy of the solution improves; however, this does not give us a clear indication of which step size to choose. The amount of improvement seems to be the same for each reduction in Δx . We solve this problem by looking at the numerical worst-case scenario—a sharp transition between uncontrolled cortex and a single electrode with proportional feedback. We then choose Δx based on its ability to resolve this sharp spatial change (Fig. 3). While this figure shows that the differences between the step sizes are still subtle, it seems that the largest one, $\Delta x = 0.448$ mm, does not provide enough detail to show the sharp transition between cortex and electrode. The smaller step sizes appear to be more accurate and provide very similar solutions. Because it will be less computationally intensive to use $\Delta x = 0.224$ mm (or $\Delta \tilde{x} = 0.0008$ dimensionless), we choose this as the step size for our simulations.

In order to verify this in the typical case with no feedback control, we plot numerical solutions with $\Delta x = 0.448, 0.224$, and 0.112 mm (Fig. 4). Because the spatial

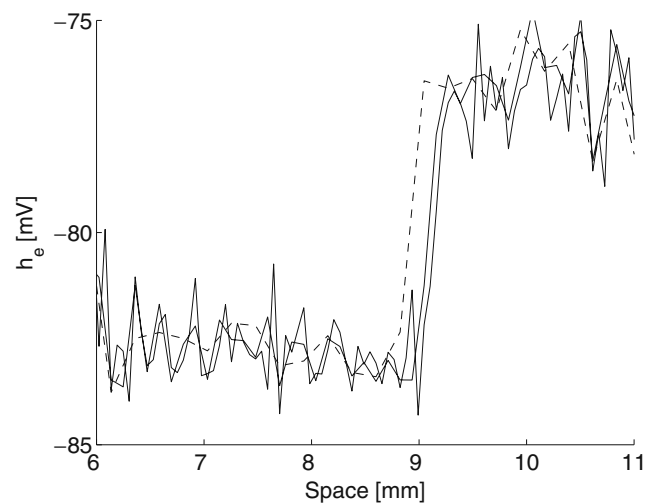


Fig. 3 Convergence of numerical solutions over an electrode nonlinearity as the spatial step size is decreased using the method from Section 3.1. The full simulation spanned 22.4 mm, but here we show only 3 mm of uncontrolled cortex and 2 mm of cortex underneath an electrode while proportional feedback is applied. This figure shows solutions for $\Delta x = 4\Delta x_0$ (dashed), $2\Delta x_0$ (solid), and Δx_0 (solid), where $\Delta x_0 = 0.112$ mm. Note that the two smallest time steps give very similar, overlapping solutions for \tilde{h}_e , while $4\Delta x_0$ gives an inaccurate result. Therefore, we choose the step size $\Delta x = 2\Delta x_0 = 0.224$ mm as a balance between accuracy and simulation cost. Other relevant parameters were $\Delta t = 2 \times 10^{-6}$ s, $N = 80,000$, $P_{ee} = 548.0$, and $\Gamma_e = 0.8 \times 10^{-3}$

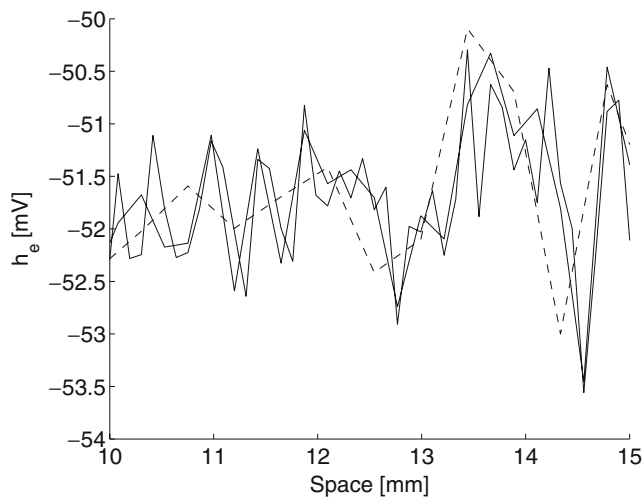


Fig. 4 Convergence of numerical solutions as the spatial step size is decreased using the method described in Section 3.1. Here we used $\Delta x_0 = 0.112$ mm, and we plotted the solutions for $\Delta x = 4\Delta x_0$ (dashed), $2\Delta x_0$ (solid), and Δx_0 (solid). The two smallest time steps give very similar results, indicating that the solution has converged; this justifies our choice of step size, $\Delta x = 0.224$ mm. This figure was created with typical excitation $P_{ee} = 11.0$ and $\Gamma_e = 1.42 \times 10^{-3}$, and $N = 80,000$ time steps were used at $\Delta t = 2 \times 10^{-6}$ s

solution appears to have converged at these step sizes, our choice of $\Delta x = 0.224$ mm is valid. The last task is to choose a value of Δt . We would like to use the largest time step for which the solution converges because this will result in the shortest computation time; this value is $\Delta \tilde{t} = 0.0001$ (dimensionless) or $\Delta t = 4 \times 10^{-6}$ s. Because we already showed that accurate solutions can be obtained with much larger values of Δt , this is an acceptable choice.

Therefore, the step sizes used in all of the following cortical simulations will be $\Delta x = 0.224$ mm and $\Delta t = 4 \times 10^{-6}$ s.

4 Model of feedback control

Our previous simulations of feedback control (Kramer et al. 2006; Lopour and Szeri 2008) utilized two key assumptions: 1) the signal measured by an electrode on the cortical surface is proportional to h_e , the mean soma potential of the excitatory neuron population, and 2) a voltage applied to the surface of the cortex via electrode directly affects the average soma voltage in that region. The first assumption allows us to define the control effort u in terms of \tilde{h}_e (in this case, \tilde{h}_e would represent the measured voltage), and the second assumption implies that the expression for u can be added directly to the SPDE model in Eq. (1). While the

latter assumption appears to be valid, there is evidence that we cannot write u as an explicit function of \tilde{h}_e as the first assumption suggests. It is likely that the voltage sensed by a surface electrode is different than the averaged soma voltage, \tilde{h}_e .

First, it is important to realize that the signal measured by an electrode is a function of the extracellular currents in the tissue, rather than the intracellular somatic potential (Nunez and Srinivasan 2006). We define the signal sensed at a point on the cortical surface to be \tilde{h}_m . Then, to understand the difference between \tilde{h}_e and \tilde{h}_m , we consider a pyramidal neuron in the cortex with one excitatory synapse as shown in Fig. 5. Say that the pyramidal neuron receives excitatory input due to a proximal synapse in layer 4 (Fig. 5(a)); this will cause intracellular flow of ions that will induce a current dipole with sources (+) on the apical dendrite near the surface and sinks (−) near the soma. The surface electrode \tilde{h}_m will sense the extracellular current source near the surface and will thus depolarize. The soma potential \tilde{h}_e will also depolarize due to the excitatory input; therefore, in this case, both \tilde{h}_e and the surface electrode show a depolarization. On the other hand, suppose that the pyramidal neuron receives excitatory input due to a distal synapse in layer 1 (Fig. 5(b)). Because the input is still excitatory, the neuron will depolarize, and this will be reflected in the soma potential \tilde{h}_e . However, this input will cause an extracellular current dipole with reverse polarity; there will be a source (+) near

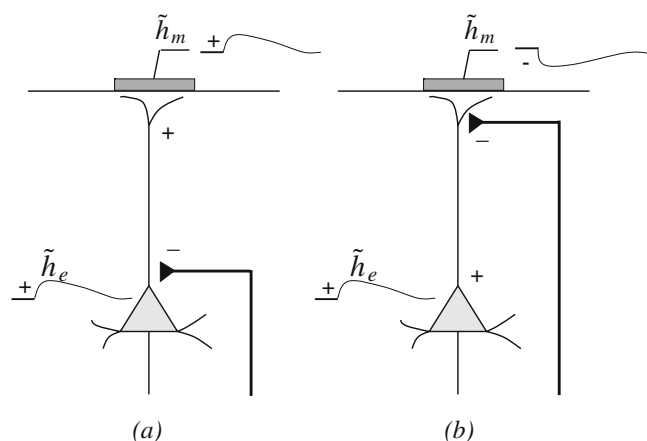


Fig. 5 Relationship of the sensed signal \tilde{h}_m to the mean soma potential \tilde{h}_e . In (a), an excitatory input synapses near the soma, causing a depolarization in \tilde{h}_e . Similarly, the dipole current generated by this synaptic event involves current sources near the cortical surface, which are manifested as a depolarization in the electrode measurement \tilde{h}_m . However, when the excitatory input occurs near the surface as in (b), the current dipole is reversed, causing opposite deflections in \tilde{h}_e and \tilde{h}_m . This figure is modeled after Box 46-1 in Kandel et al. (2000)

the soma and current sinks (–) near the surface. This means that the voltage sensed by the surface electrode will show a hyperpolarization. Therefore, in this case, the deflection of \tilde{h}_e and the signal seen by the surface electrode are different (Kandel et al. 2000).

This implies that we should no longer use \tilde{h}_e as the measured electrical potential in our expression for the control effort u . Instead, the measurement will be a function of the currents in the cortex due to synaptic inputs, denoted as \tilde{h}_m . We will refer to this as the sensed signal. Then, for the purposes of feedback control, the applied effort u will be a function of \tilde{h}_m .

4.1 Basic form of model

To determine the composition of the sensed signal, we must consider the extracellular current flows due to three types of synaptic input (local intracortical input, long-range corticocortical input, and subcortical input). To do this, we need to know whether the inputs are excitatory or inhibitory and whether they synapse near the surface or near the soma. In what follows, we take care to distinguish depolarization in the electrode measurement from depolarization of the soma.

- **Local intracortical inputs, $N_e^\beta \tilde{S}_e$ and $N_i^\beta \tilde{S}_i$.** Within a cortical macrocolumn, excitatory synapses tend to occur close to the surface, while inhibitory synapses are located near the soma (Nunez and Srinivasan 2006; Spruston 2008). Thus, the excitatory inputs are as depicted in Fig. 5(b), and the inhibitory inputs have the geometry of Fig. 5(a) but with the opposite sign (because Fig. 5 depicts *excitatory* inputs). Each of these configurations will cause a hyperpolarization in the electrode measurement; therefore, both terms will have negative signs in the measurement model: $-AN_e^\beta \tilde{S}_e - BN_i^\beta \tilde{S}_i$, where A and B are positive constant weights to be determined.
- **Long-range corticocortical input, $\tilde{\phi}_e$.** Corticocortical inputs are exclusively excitatory (Steyn-Ross et al. 2003; Liley et al. 2002; Nunez and Srinivasan 2006) and tend to synapse near the surface (Nunez and Srinivasan 2006; Spruston 2008). More specifically, layers 2 and 3 of the cortex seem to have a higher density of corticocortical inputs (Kandel et al. 2000; Nieuwenhuys 1994). As shown in Fig. 5(b), this input type will cause a hyperpolarization in the electrode signal and will thus have a negative sign in the measurement model: $-C\tilde{\phi}_e$, where C is a positive constant weighting factor to be determined.
- **Subcortical inputs, $(P_{ee} + \tilde{\Gamma}_1)$ and $(P_{ie} + \tilde{\Gamma}_3)$.** While the distribution of these synapses is not clear-cut, it seems that subcortical inputs terminate most

densely in layer 4 near the soma (Kandel et al. 2000; Nieuwenhuys 1994). Because P_{ee} is an excitatory input of the type shown in Fig. 5(a), it will have a depolarizing effect on the electrode measurement; therefore, we give it a positive sign: $+D(P_{ee} + \tilde{\Gamma}_1)$, where D is a positive constant weight. On the other hand, P_{ie} is inhibitory and will thus have a hyperpolarizing effect: $-E(P_{ie} + \tilde{\Gamma}_3)$, where E is a constant weighting factor. The values of D and E are to be determined.

Incorporating all three types of synaptic input gives us this basic expression:

$$\text{measured current} \sim -AN_e^\beta \tilde{S}_e - BN_i^\beta \tilde{S}_i - C\tilde{\phi}_e + D(P_{ee} + \tilde{\Gamma}_1) - E(P_{ie} + \tilde{\Gamma}_3), \quad (15)$$

with A, B, C, D, E of positive sign but (so far) unknown magnitude. The consequence of these inputs is only evident after synaptic transmission. Therefore, we include a rate constant for this process by using an equation similar to that of \tilde{I}_{ee} in the SPDE model. Let \tilde{I}_m represent the current measured at the cortical surface and T_m represent a rate constant. Then

$$\left(\frac{1}{T_m} \frac{\partial}{\partial t} + 1\right) \tilde{I}_m = F(-AN_e^\beta \tilde{S}_e - BN_i^\beta \tilde{S}_i - C\tilde{\phi}_e + D(P_{ee} + \tilde{\Gamma}_1) - E(P_{ie} + \tilde{\Gamma}_3)), \quad (16)$$

where A, B, C, D, E , and F are positive constant weights. We choose $T_m = 12.0$ to match the rate constant of the excitatory population T_e . The values of A through E will depend on the number of synapses of each type and the average distance of the synapse from the soma. The coefficient F is a gain parameter that will scale the magnitude of all the synaptic inputs; this ensures that they have the appropriate amount of influence over the electrode measurement \tilde{I}_m . In addition, we can think of F as containing the effective resistance of the cortex. Recall that the electrode measurement is determined by *currents* in the cortex, yet the components on the right side of (16) are based on voltages. Because the currents produced by these voltages can be calculated with Ohm's Law (Kandel et al. 2000), the gain parameter F provides the necessary conversion.

To complete the model of the electrode measurement \tilde{h}_m , we must account for the reversal potential of the cortical neurons. This determines the direction of current flow associated with the inputs described above (we previously assumed that the neurons were

Table 2 Values of the coefficients for Eq. (16)

Coefficient	Input type	Lopour/Szeri value	Liley/Wright value
<i>A</i>	Local excitatory	0.413	0.324
<i>B</i>	Local inhibitory	0.092	0.088
<i>C</i>	Long-range excitatory	0.458	0.583
<i>D</i>	Thalamocortical excitatory	0.034	0.006
<i>E</i>	Subcortical inhibitory	0.004	0.000

The first column of values was estimated as described here, while the second column was derived using probabilistic methods in Liley and Wright (1994)

at resting potential). We once again take our cue from the SPDE model and define

$$\tilde{h}_m \equiv (h_0^e - \tilde{h}_e) \tilde{I}_m = \left(\frac{45 - h_e}{-70} \right) \tilde{I}_m. \quad (17)$$

Thus, (16) and (17) comprise a complete model of the potential sensed by a cortical surface electrode, \tilde{h}_m . In our simulations of feedback control, the applied electric field u will be a function of this variable.

4.2 Estimation of coefficients

We have not yet assigned numerical values to the coefficients *A* through *E*. To do this, we think of them as the percentage of pyramidal neuron synapses due to each source. For example, *A* will represent the percentage of synapses on any given pyramidal neuron that come from other excitatory neurons in the same macrocolumn. There are three physiological relationships that allow us to determine these values:

1. The number of synapses on pyramidal cells due to local cortical neurons is roughly equal to the number of synapses due to cortical neurons in other macrocolumns or in the contralateral hemisphere (Abeles 1991). This implies that $A + B = C$.
2. Approximately 98 percent of synapses on pyramidal cells are corticocortical, while 2 percent are thalamocortical (Abeles 1991; Nunez and Srinivasan 2006). This implies that $A + B + C = .98$ and $D + E = .02$.
3. Roughly 90 percent of all cortical synapses are excitatory and 10 percent are inhibitory (Abeles 1991; Braitenberg and Schüz 1998). This implies that $A = 9B$ and $D = 9E$.

After solving the above equations, we account for the fact that synapses near the soma will have a greater influence on the electrode measurement (Nunez and Srinivasan 2006) by multiplying *B*, *D*, and *E* by a factor of two.

While this method of estimation may seem crude, others have achieved similar results through more detailed probabilistic analysis (Liley and Wright 1994).

Both are listed in Table 2 for comparison; note that each set of coefficients has been scaled to add to 1. In simulation, when comparing the two sets of coefficients *A* through *E*, the only difference in the resulting \tilde{h}_m appears to be an offset. Because offsets are not reflected in EEG measurements, this difference is inconsequential. Hence, two completely independent approaches provide essentially equivalent coefficients for the sensed signal.

4.3 Verification of full model

We can verify our model by returning to the hypothetical pyramidal neuron described at the beginning of

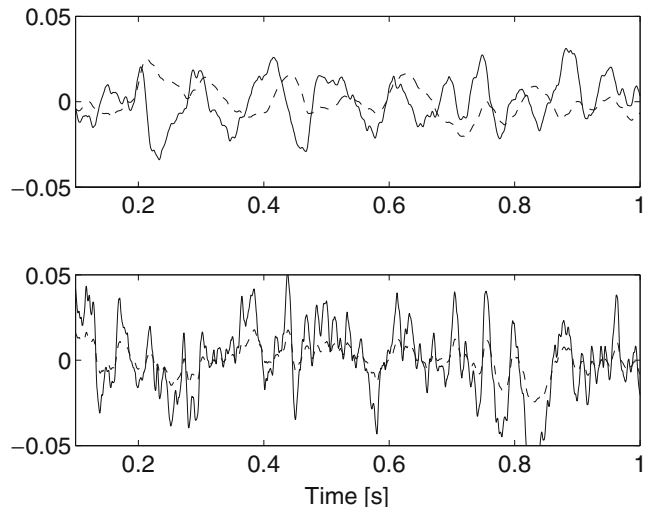


Fig. 6 Comparison of \tilde{h}_m (solid) and \tilde{h}_e (dashed) with two sets of parameters. At typical levels of excitation where $P_{ee} = 11.0$ and $\Gamma_e = 1.42 \times 10^{-3}$, the two signals are negatively correlated (top subfigure), as predicted by our physiological analysis. This negative correlation is especially noticeable if the synaptic time delay is taken into account by shifting \tilde{h}_e slightly to the left. However, when we simulate a strong excitatory input near the soma by setting $P_{ee} = 1,000.0$, the signals become positively correlated (bottom subfigure). This was also predicted by the physiology of the cortex, and thus helps justify our model of \tilde{h}_m . Note that for both sets of parameters, the signal offsets (means) were removed to facilitate a direct comparison

Section 4 and based on Box 46-1 in (Kandel et al. 2000). As before, say that we are modeling the electrode measurement of a pyramidal neuron with excitatory inputs in both cortical layers 1 and 4. If a majority of the inputs occur in layer 1 near the cortical surface as in Fig. 5(b), \tilde{h}_e and \tilde{h}_m will have similar dynamics, but a hyperpolarization in one signal will be a depolarization in the other; the signals will be negatively correlated. This behavior is seen in our model for \tilde{h}_m . If we run a simulation with the typical set of parameters (where local and corticocortical connections dominate because P_{ee} is low), we see that \tilde{h}_e and \tilde{h}_m are negatively correlated (top of Fig. 6). On the other hand, when the strongest input is near the soma in layer 4 as in Fig. 5(a), it will have the same effect on both signals, and they will be positively correlated. This, too, is demonstrated by the measurement model. We can simulate a large excitatory input near the soma by increasing the value of P_{ee} ; when we run the simulation with this change, we see that \tilde{h}_e and \tilde{h}_m become positively correlated (bottom of Fig. 6). Thus, the measurement model accurately reproduces the physiological effects of varying cortical inputs, and we will use it in subsequent simulations of feedback control. The applied electric field u will be a function of \tilde{h}_m as opposed to \tilde{h}_e .

For reference, we compare \tilde{h}_e and \tilde{h}_m at seizure parameters ($\Gamma_e = 0.8 \times 10^{-3}$ and $P_{ee} = 548.0$) in Fig. 7. In

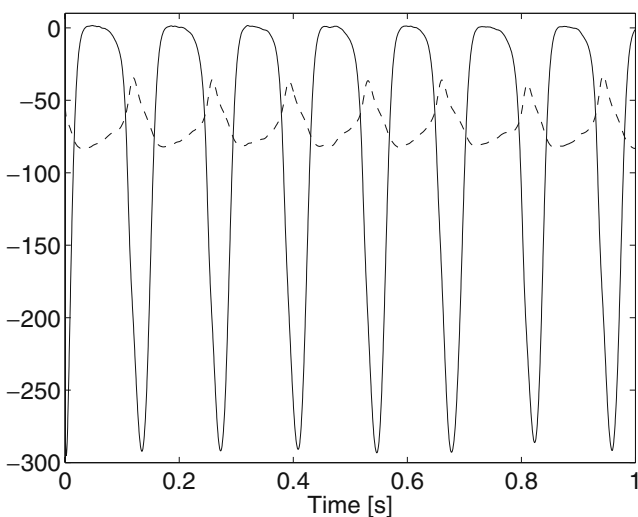


Fig. 7 Comparison of the electrode measurement h_m (mV, solid) and the mean soma potential h_e (mV, dashed) at levels of sub-cortical excitation that cause seizure-like oscillations, $P_{ee} = 548.0$ and $\Gamma_e = 0.8 \times 10^{-3}$. Here we see that the two signals are negatively correlated, but have very similar dynamics. This indicates that it should be possible to perform feedback control using our new measurement model \tilde{h}_m (as we did previously with \tilde{h}_e in Kramer et al. 2006), although a positive gain may be necessary

this case, the signals have a large negative correlation, but very similar dynamics. This suggests that it will be possible to suppress seizures with feedback control based on \tilde{h}_m (as it was with \tilde{h}_e in Kramer et al. 2006), although we may need to use a gain of the opposite sign.

5 Simulation of integral control

In choosing a function to represent the applied electric field u , we start with the concept of proportional feedback control. This is the simplest and most common type of control—intuitively, the applied effort should be proportional to the error between the measured signal and its desired value. In this case, we can define proportional control as

$$u = a(x, t)(\tilde{h}_m + b), \quad (18)$$

where $a(x, t)$ is the control gain and \tilde{h}_m is the measured electrode potential; it is calculated by taking the spatial average of \tilde{h}_m (17) under each electrode. The parameter b is a constant offset that can be tuned to achieve the desired equilibrium value of \tilde{h}_e . While this control algorithm is able to suppress the seizure-like oscillations of the model, it would be difficult to implement safely. Whenever stimulation is applied, it is important that the process be chemically reversible in order to prevent damage due to the production of new chemical species. There is a threshold for reversibility called the “reversible charge injection limit,” which signifies the maximum allowable charge injection before the polarity is reversed (Robblee and Rose 1990). Because a proportional controller does not penalize the amount of effort used (i.e. the magnitude of the applied electric field), it relies on large signals of only one sign, which would exceed this threshold over time. The chemical processes associated with this type of stimulation would therefore be irreversible and damaging to cortical tissue. The simulation results based on this type of control have been presented in previous publications (Kramer et al. 2006), and we do not repeat them here.

To improve on this method, we may consider adding a derivative or integral component to the controller, or even using all three terms to create a proportional-integral-differential (PID) controller (Franklin et al. 2002). The derivative term increases or decreases the control effort based on the rate of change of the error. This can reduce the response time of the controller because the derivative term “anticipates” the behavior of the system. Simulation results with a PD controller were presented in Kramer et al. (2006).

Because the differential controller utilizes the same harmful voltages mentioned previously, we choose to implement a controller with an integral term:

$$u = a(x, t)(\tilde{h}_m + b) + c(x, t) \int u \, dt. \quad (19)$$

Here, $c(x, t)$ is another gain term. It will be negative, meaning that this new term will oppose the total integral of the applied voltage u . If the integral of u is positive, it will add a negative component to the applied voltage, and vice versa; in this way, it pushes the integral of u to zero. In other words, it forces the applied signal to be charge-balanced and thus safe for cortical tissue. Because we have also included the proportional control term, this feedback setup will still suppress the seizure oscillations. Note that this is different than traditional integral control, which is based on the integral of the *error* between the desired value of the signal and its actual value.

In addition to adding the integral term and using the new measurement model \tilde{h}_m , we incorporate the step sizes determined in Section 3. Because we are using smaller increments of space, we can make one further improvement: we add a more detailed electrode profile to the feedback simulation. We previously assumed that the electrodes maintained a constant profile across their surface while measuring or applying the stimulus (i.e. every point on the electrode sensed or provided the same value) and that no tissue beyond the edge of the electrode was affected by its activity. However it has been shown experimentally that this is not the

case (Suesserman et al. 1991). Here, we take the first step towards a realistic electrode model by including a smooth falloff at the electrode edges. The falloff is incorporated into the function $a(x, t)$, so that the gain varies between zero (over uncontrolled cortex) and a_{\max} (under the electrodes) via a hyperbolic tangent function. This gain function is used in the application of control to indicate that the influence of the electrode decreases with distance, and it is also applied during sensing with $a_{\max} = 1$ to indicate that the cortical tissue has less impact on the electrode measurement as distance increases. We have not yet included any variation over the surface of the electrode, but this is certainly an adjustment that can be considered in the future.

With this approach, we are in a position to simulate the suppression of seizure waves using total integral feedback control as defined in (19). The results shown here were generated using an Intel Core 2 Duo 2.13 GHz processor, and the calculations took roughly 5 min at the highest spatial and temporal resolutions. The code was written and executed in MATLAB and has been provided as supplementary online material (Online Resource 1).

Figure 8 shows an example of the seizure-like oscillations, represented by the mean soma potential of the excitatory population h_e as it varies in space and time. More specifically, we have simulated an uncontrolled strip of cortex 200 mm long over 0.5 s, and the value of h_e [mV] is represented by grayscale. The seizure waves occur due to our choice of $\Gamma_e = 0.8 \times 10^{-3}$ and a Gaussian P_{ee} distribution with a maximum value

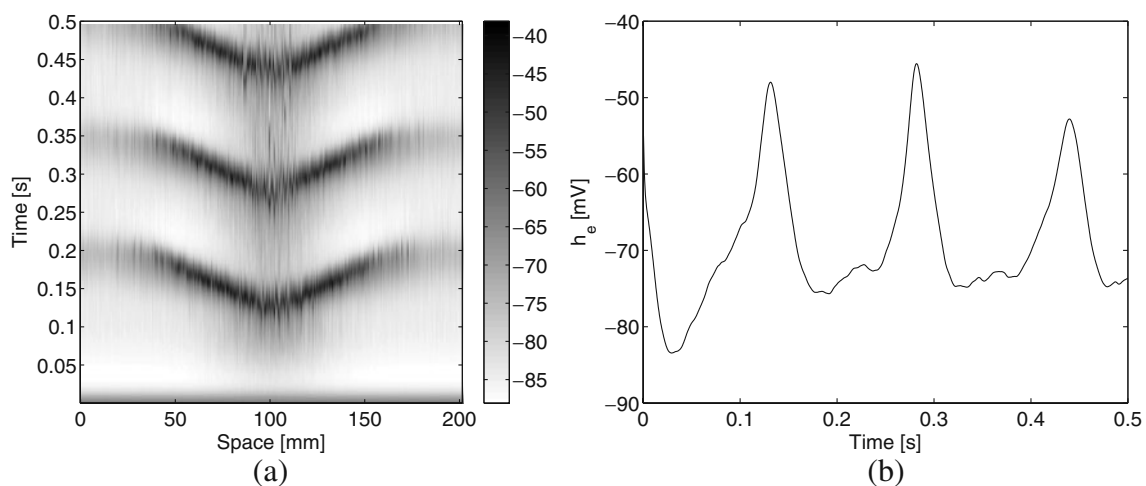


Fig. 8 (a) Seizure waves traveling on the simulated cortex, with parameters $P_{ee} = 548.0$ and $\Gamma_e = 0.8 \times 10^{-3}$. The characteristics of the wave are determined by the distribution of P_{ee} ; here it is a Gaussian curve, so the wave starts in the center where P_{ee} is

at its maximum (548.0) and propagates outward until the level of excitation is too low to support it. In this case, no feedback control is applied, so the waves will reoccur indefinitely. (b) Plot of h_e in time extracted from (a) at 100.8 mm

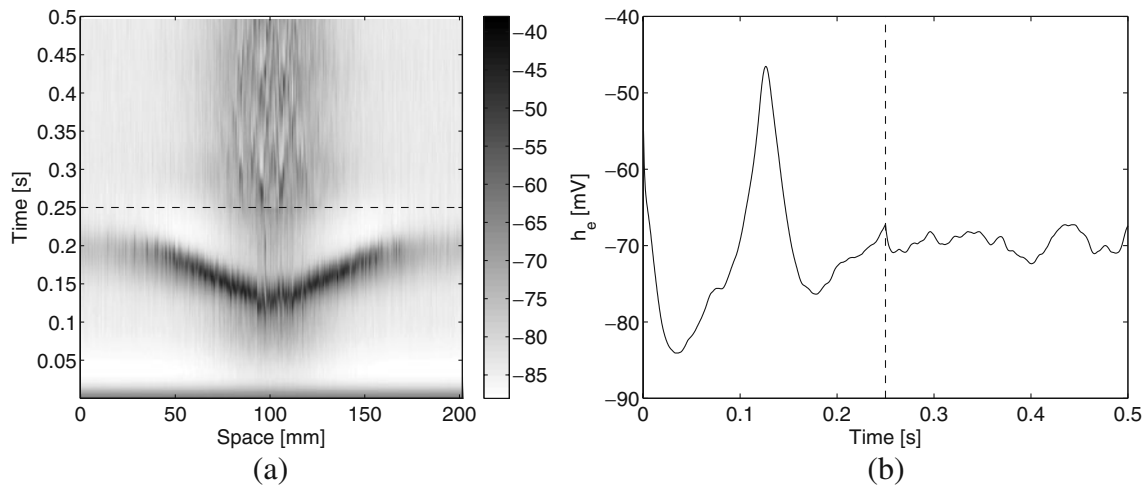


Fig. 9 (a) Plot of h_e [mV] in space and time. In the first 0.25 s, we see a seizure wave traveling on the simulated cortex because we have set the excitation parameters to $P_{ee} = 548.0$ and $\Gamma_e = 0.8 \times 10^{-3}$. At 0.25 s (dashed line), the integral controller (19) is turned on, and we see that the seizure-like waves are immediately

suppressed. For this simulation, we used \tilde{h}_m as the electrode measurement, and control was applied via five electrodes of width 11.2 mm with a profile defined by the hyperbolic tangent function. The controller gains were $a_{\max} = 8$, $b = -0.1$, and $c = -8$. (b) Plot of h_e in time extracted from (a) at 100.8 mm

of 548.0. They spontaneously arise in locations with sufficiently high P_{ee} (here, this “hot spot” is at $x = 100$ mm) and travel outward until the level of excitation is too low to support them. This is why the waves terminate before they reach the edges of the simulation space.

Figure 9 shows the effect of total integral feedback control on this seizure-like behavior. In the first half

of the time interval, we see a seizure wave emanating from the center of the space; then, at 0.25 s the control is switched on, and the pathological seizure activity quickly disappears. In this case, the feedback stimulation is applied via five electrodes that are 11.2 mm across. The geometry of these electrodes can be clearly seen in Fig. 10, which shows the value of the applied signal u [mV] for the same simulation. Here, we see

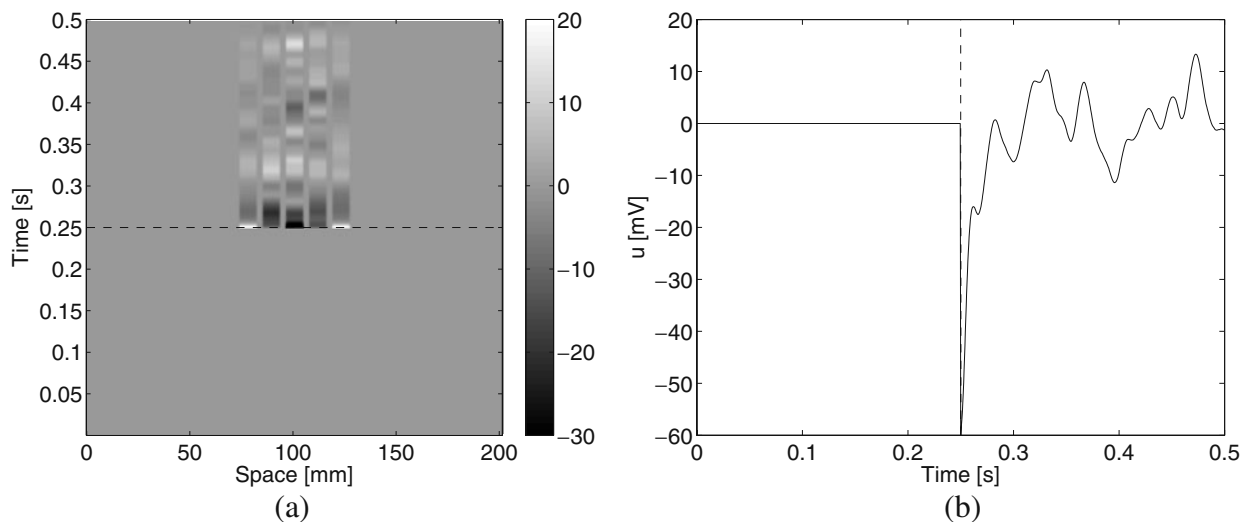


Fig. 10 (a) Plot of applied effort u [mV] in space and time, corresponding to the simulation of feedback control in Fig. 9. This shows that the potential applied by the electrodes is zero until $t = 0.25$ s, when the controller is turned on (dashed line). Then each electrode oscillates between positive and negative signals, indicative of the “charge-balanced” nature of the control.

For this simulation, the average magnitude of total applied effort over all five electrodes was 0.23 mV. This value will approach zero as t increases, and it can also be reduced by increasing the integral control gain, c . Note that, for clarity, the grayscale in this figure is the opposite of the ones used in Figs. 8 and 9. (b) Stimulation applied by the center electrode in (a)

that the applied voltage is zero for the first half of the time interval, and then each electrode varies between positive and negative voltages once the controller is turned on. While the largest potential applied by a single electrode is roughly 60 mV, the total signal applied by each electrode is very close to zero, due to our choice of control law u . This is the desired result because it indicates that each electrode applies a balanced signal. Lastly, note that it is not necessary to have electrodes covering the entire length of the seizure wave. When we place the electrodes at the center of the “hot spot,” they are able to halt the outward motion of the wave. For more information on the parameters used to create the figures, please refer to the figure captions.

Thus, in Figs. 9 and 10 we have demonstrated that the new model for electrode measurements \tilde{h}_m can be used to suppress seizure waves via feedback control. If the controller u contains the integral of the total applied effort, then this can be done in a manner that is thought to be safe for cortical tissue.

6 Discussion

Here we have presented several novel approaches to exploration of a model of feedback control for epileptic seizures in humans. We first verified the strong convergence of numerical solutions to the model of the cortex, paying special attention to discontinuities that may occur at electrode edges. This allowed us to choose appropriate step sizes for our simulations; because the spatial step size Δx was small relative to the size of the electrode, we were able to incorporate a more detailed electrode profile into the simulation. Then, based on evidence that the mean soma potential \tilde{h}_e cannot be used as the measurement for feedback control, we developed a new model \tilde{h}_m to represent the measurement of cortical surface electrodes. This model was based on the currents flowing in the cortex and was used for all simulations of feedback control. Those simulations utilized a new control algorithm containing the total integral of the applied potential u . Not only did this succeed in suppressing the seizure-like oscillations, but it guaranteed that the applied signal would be charge-balanced and therefore safe for cortical tissue.

Of course, there are always improvements to be made. In this work, we have assumed that each electrode can be simultaneously sensing and applying the control signal. This is not realistic; ideally, we would model separate electrodes for these two tasks, with the geometric properties chosen to match existing experimental setups. Also, as mentioned previously, it would be possible for us to improve the electrode

profile used in simulation. Here, we incorporated a simple profile to demonstrate our capability to do so, but it would be more accurate to base our choice on existing theories of the potential difference across an electrode surface (Rubinstein et al. 1987). It may also be possible to account for electrochemical changes that occur in the vicinity. Lastly, we note that our use of the phrase “charge-balanced” should be taken lightly. Our controller measures and applies a *voltage*, so the integral term pushes the total voltage (over time) towards zero. Although, in concept, this is similar to having a charge-balanced signal, it does not guarantee that the applied charges will be balanced and safe. This could be remedied by utilizing a controller that measures cortical potential and applies a *current*. Not only would this be more accurate, but it would facilitate future comparisons with experiments, most of which are done in this manner (Colpan et al. 2007; Sunderam et al. 2006).

Validation via experimentation is only one of the many possible future directions of this work. For example, we could extend our model to two dimensions and use it to study seizure waves, or we could simulate experimental phenomena such as irregular, spiral, and plane cortical waves (Schiff et al. 2007). Related theoretical work has suggested that pre-processing of data using a Kalman filter can provide greater flexibility in the control of waves while minimizing the amount of energy needed to do so (Schiff and Sauer 2008). This concept could be readily applied to the simulations discussed here. Another possible avenue of future work is the investigation of spatial properties of our feedback model. We have the capability to do simulations with any number of electrodes at any size and spacing, which is a luxury not afforded to experimentalists. Theoretical work in this area may provide insight into important questions such as: how does the size of a seizure relate to the number and sizes of the electrodes needed to control it successfully? Where should the electrodes be placed for maximum effectiveness? What are the necessary resolutions for sensing and actuation? It is our hope that this work will act as a stepping stone to such intriguing questions.

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Chapter 86

Spatial Considerations of Feedback Control for the Suppression of Epileptic Seizures

Beth A. Lopour and Andrew J. Szeri

Abstract Control of epileptic seizures through brain electrical stimulation is currently the focus of experimental and theoretical research. Here we use a mean-field model of the human cortex to study spatial issues related to two types of feedback control. We start with a 14th order set of stochastic partial differential equations from Kramer et al. (Journal of Computational Neuroscience 22:63–80, 2007), add either proportional or differential control, and then simulate the effect on seizure-like behavior. We analyze the results based on changes in electrode size, electrode spacing, and the number of electrodes in the grid.

Background and Model

There are millions of people in the US that are affected by epileptic seizures, and roughly one in five do not respond to medication [1]. For those that do not respond to anti-epileptic drugs, the available treatments are invasive and dangerous. It may even be necessary to surgically resect the seizing portion of the cortex. During pre-surgical evaluations, doctors found that applying pulses of electrical stimulation to the cortex can suppress epileptiform afterdischarges, which are often precursors to full-blown seizures [2]. In addition, it has been found that sinusoidal voltages can be applied to various brain structures and nerves to reduce the severity of epileptic seizures [3, 4]. Going one step further, some researchers have experimented with the application of electric fields via feedback control [5]. Currently, the mechanism by which brain electrical stimulation works is unknown, and no *optimal* techniques have been developed. However, research in this area may eventually lead to implanted devices that can automatically stop seizures when they occur, or perhaps even prevent them.

Because experiments in electrical stimulation can be dangerous for humans, mathematical models are useful for exploring these ideas. Here, we utilize a

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mesoscale stochastic PDE model of the cortex that was first developed in the 1970's [6] and has since undergone revisions and improvements [7]. It has not only been used to study epilepsy [8, 9], but also anesthesia [10] and sleep [11]. The 14th order model consists of dimensionless partial differential equations in space (x) and time (t). The equations contain four stochastic terms, which represent subcortical inputs and are approximated by Gaussian-distributed white noise (scaled in space and time), in addition to several nonlinear terms. We will incorporate two different feedback controllers and analyze their utility for stopping epileptic seizures.

While the model contains a large number of states and parameters, there are only three that are central to the results described here: (1) \mathbf{h}_e – a variable that represents the mean soma voltage of the neurons in the model and is assumed to be proportional to the voltage that can be measured via EEG. Because it is related to an observable quantity, h_e is the variable that will be used for feedback control. (2) Γ_e – a parameter that represents the influence of excitatory input on the averaged soma voltage. (3) \mathbf{P}_{ee} – a parameter that represents the subcortical input to the excitatory neurons in the cortex. For a full statement of the model, the reader is directed to [9].

Simulations and Results

At typical values of the parameters described above, $\Gamma_e = 1.42 \times 10^{-3}$ and $P_{ee} = 11.0$, the simulated cortical voltage resembles the fluctuations of a typical EEG signal. The variation in voltage is roughly 8 mV centered around -52 mV. However, if we change the values of Γ_e and P_{ee} we observe remarkably different behavior. In a “hyper-excited” state, with $\Gamma_e = 0.8 \times 10^{-3}$ and $P_{ee} = 550.0$, we see repeated large-amplitude oscillations in the model that are reminiscent of a seizure. The variation in voltage increases to roughly 50 mV centered around -60 mV. Because we believe h_e to be related to the electric fields measured and induced by electrodes on the cortex, this variable is a prime candidate for feedback control. The goal is to use feedback to stop seizures that are in progress and move the cortex back to its normal operating range.

In order to implement this type of control, we need to measure the electric field on the cortex through a set of electrodes, use those values to calculate the control, and then apply the control voltages to the brain via a different set of electrodes. There are several assumptions we make in order to simulate this process. First, we assume that h_e is directly proportional to the EEG measurement, so we may use h_e to calculate the control [12]. Next, we ignore the spread of the electric field beyond the electrode edges. Lastly, we assume that the output of the electrode adds directly to the electric field of the neuron itself; in the model, this means that we can simply add the control law to $\partial h_e / \partial t$. With these in mind, we model *proportional control* by adding an $a(x, t)\tilde{h}_e$ term to the first equation of the model:

$$\frac{\partial \tilde{h}_e}{\partial t} = 1 - \tilde{h}_e + \cdots + a(x, t)\tilde{h}_e. \quad (86.1)$$

Here, $a(x, t)$ represents the gain, which can vary across the cortex, and \bar{h}_e is a local spatial average of \tilde{h}_e . The number of points in the average is equal to the size of the simulated electrode. Also, by setting the gain to zero in strategic locations, we can delineate individual electrodes. These parameters allow analysis of the effects of changing the electrode size, electrode spacing, and number of electrodes.

Alternatively, we could add the term $a(x, t)(\tilde{h}_e[t] - \tilde{h}_e[t - \tau])$ to model *differential control*. As before, $a(x, t)$ represents the discrete controller gain, but we have now incorporated a delay time τ . This method of control has been shown to stop seizure-like behavior in the model [8], with the added benefit of using voltages that are less likely to damage brain tissue.

Because the simulation parameters present a seemingly infinite number of possible combinations, we have focused on three key issues:

1. **Number of electrodes:** This defines the number of electrodes placed side-by-side on the cortex, similar to electrode grids used in experiments; the gain $a(x, t)$ is set to zero for all spatial steps outside the grid.
2. **Electrode size:** To represent electrodes covering multiple spatial points, we determine the average of h_e over the points and use it to calculate one control voltage. We then apply that voltage to all of the points covered by the electrode.
3. **Spacing between electrodes:** To simulate locations unaffected by the applied electric field, the gain $a(x, t)$ is set to zero for all points determined to be between electrodes.

In this analysis, we do not address the issue of sensing and controlling at different cortical locations, and we have assumed that the electric fields applied by each electrode are constant in space. For simplicity, we neglect extracellular electrochemical changes in the immediate vicinity of the electrodes.

We simulated feedback control of the stochastic PDE's using a predictor-corrector algorithm written in Matlab. The stochastic terms were incorporated using the Euler-Maruyama method, which ensures that the quality of the noise remains the same regardless of the integration step sizes. While more sophisticated solvers would provide better convergence and stability properties, this method was chosen due to the complexity of the system.

The following parameters were used in all simulations: the gain $a(x, t) = 0$ for $0 \leq t \leq 0.5$; for $t > 0.5$ within the electrode grid, $a(x, t) = -2$ for proportional control, and $a(x, t) = -10$ with a delay of $\tau = 20$ ms for differential control; the integration was performed over 10,000 time steps with a step size of 0.1 ms; 50 spatial steps were used with a step size of 7 mm; $\Gamma_e = 0.8 \times 10^{-3}$; P_{ee} was normally distributed in space with a maximum of 550; the magnitude of the stochastic terms was $\alpha = 16$. These parameter choices were based on those in [8].

Proportional Control Results

A typical simulation of proportional control is shown in Figure 86.1a shows the resulting cortical voltage h_e when control is applied to four 35 mm electrodes with

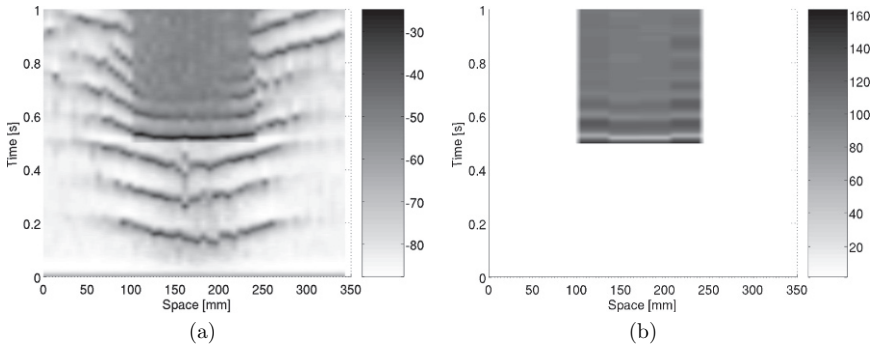


Fig. 86.1 Simulation of proportional control. (a) h_e output for four 35 mm electrodes with 0 mm spacing. The strong wavelike solutions emanating from the center of hyper-excitation for $t < 0.5$ are those associated with a seizure. (b) Control effort for the same parameters as in (a)

0 mm spacing. Figure 86.1b depicts the voltage applied to the cortex by the controller. In the locations where electrodes have been applied, the proportional control succeeds in stopping the seizure-like oscillations; note that when the controller is turned on at $t = 0.5$, h_e quickly moves to a steady voltage of roughly -50 mV. This result can be obtained with electrodes of *any* size. However, there are several aspects of this control that are not optimal. While the controller works well wherever electrodes are applied, new large-amplitude waves appear to emanate from the controlled area. This seems to occur regardless of the number of electrodes in the grid. Also, the required control effort consists of a large positive voltage, which could potentially damage cortical tissue.

Simulations with non-zero spacing mimic the results presented here. Proportional control is effective at stopping seizure-like oscillations at the electrodes, but does not necessarily prevent waves from occurring in the gaps. Hence, the electrode size is not as critical as the length of spacing between them.

Differential Control Results

Figure 86.2 shows a simulation of differential control. As with proportional control, Fig. 86.2a shows the resulting h_e voltage while Fig. 86.2b displays the calculated control effort. The simulation was done with six 14 mm electrodes at a spacing of 0 mm. In the ideal case, with an electrode size of 7 mm (one nondimensional step) and 0 mm spacing, the controller completely halts the large-amplitude oscillations and brings the cortical voltage to roughly -70 mV. However, with larger electrodes, we see that small sections of high voltage persist even after the controller is turned on. This effect is enhanced as electrode size and spacing are increased – more and more segments of the wavefront remain until the controller is not capable of stopping the seizure-like oscillations. Also, note that the equilibrium value of -70 mV does not match the typical mean soma potential associated with $\Gamma_e = 1.42 \times 10^{-3}$ and $P_{ee} = 11.0$. On the other hand, differential control does offer several advantages

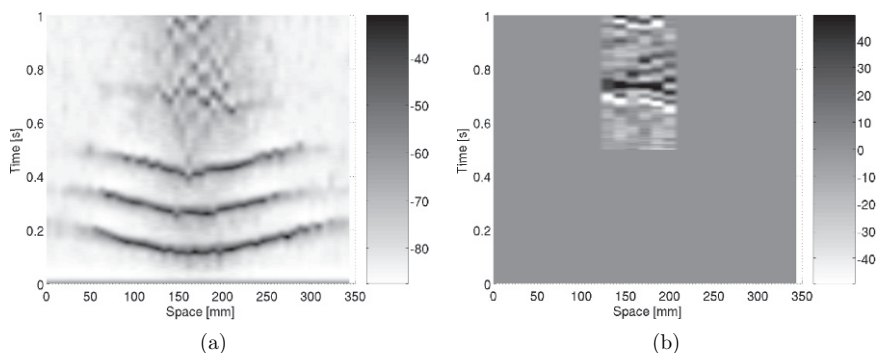


Fig. 86.2 Simulation of differential control. (a) h_e output for six 14 mm electrodes with 0 mm spacing. (b) Control effort for the same parameters

over proportional control. First, despite the fact that the grid is smaller than the extent of the traveling waves, the controller is able to stop the seizure-like oscillations. No new waves develop at the edge of the electrode grid. Second, the control effort used in this case is lower amplitude and oscillates between positive and negative values. This type of signal is much safer for cortical tissue, as discussed in [8] and the references therein.

Discussion

Overall, both proportional and differential control have advantages and disadvantages. Proportional control is effective regardless of electrode size, and it also returns h_e to its typical value of roughly -50 mV. However, the large positive voltages it uses could potentially be damaging to cortical tissue, and it sometimes causes new oscillations to arise. Alternatively, differential control uses safer voltages and requires smaller grid sizes, but its effectiveness quickly deteriorates with increasing electrode size and spacing. It also has an equilibrium value of approximately -70 mV, which does not match the typical cortical state.

Some of these results may be connected to neuro-physiological properties. For example, the degree of correlation between spatial points on the cortex is related to P_{ee} ; this relationship may give insight into the size of electrode needed to stop a seizure. Future work will investigate such ideas.

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